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East Hartford, Connecticut 06108

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Final Study Report of an Applications Study of Advanced Power Generation Systems Utilizing Coal-Derived Fuels CR-161692

Volume II - Technical Report

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Date March 1981

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Final Study Report of an Applications Study of Advanced Power Generation Systems Utilizing Coal-Derived Fuels

Volume II - Technical Report

NAS8-33996

DPD No. 593

DR No. 6

Fred. L. Robson / Chief, Utility Power Systems

DATE March 1981

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FOREWORD

This application study of advanced power generation systems utilizing coal-derived fuels was performed by United Technologies Research Center with subcontract assistance from Burns and Roe, Inc. The study program has been prepared under Contract NAS8-33996 for Marshall Spare Flight Center of the National Aeronautics and Space Administration. The technical work described herein was initiated in September 1980 and completed in March 1981.

The technical results of this application study are presented in two parts. The first part is an Executive Summary (Volume I) which provides a concise review of all major elements of the study. The second part is a detailed Technical Report (Volume II) describing the technology status of the three advanced power generation systems studied (combined cycle gas turbine, fuel cells, and magnetohydrodynamics) and the performance of these generation systems either utilizing a medium-Btu coal-derived fuel supplied via pipeline from a large central coal gasification facility or integrated with a gasification facility for supplying medium-Btu coal-derived fuel gas.

The United Technologies Research Center provided overall program management and the major contributions to the combined cycle gas turbine and fuel cell power plant evaluation. This effort involved:

Fred L. Robson, Program Manager Robert D. Lessard, Deputy Program Manager William A. Blecher Stewart J. Lehman W. Richard Davison

Burns and Roe provided the major contributions to the magnetohydrodynamics power plant evaluation. This effort involved:

Albert W. Carlson

The study program was conducted under the general direction of Richard D. Kramer and Robert Giudici of Marshall Space Flight Center.

The support and assistance of all the aforementioned individuals and organizations in contributing to the successful completion of this study program is acknowledged with sincere appreciation.

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Final Study Report of an Applications Study of Advanced Power Generation Systems Utilizing Coal-Derived Fuels

Volume II - Technical Report

One of the most promising methods for utilizing coal for electric power generation is to gasify it and then combust the fuel gas in an advanced technology power generation system. The analysis of gasification/power generation systems has been the subject of numerous studies in recent years. Despite the proliferation of studies it is often difficult to effectively utilize the data generated because of the differing guidelines which have existed among the studies. Consequently, if data were desired for a specific generation system application, it is often necessary to develop the data unique to that application.

Two applications of interest to Marshall Space Flight Center (MSFC) involve power generation systems that either 1) utilize a medium-Btu coal-derived fuel gas supplied via pipeline from a large central coal gasification facility, or 2) are integrated with a gasification facility for supplying medium-Btu coal-derived fuel gas. Although previous studies have considered these applications, the requirement still exists to evaluate the employment of specific power generation systems in these applications, namely, combined-cycle gas turbines, fuel cells, and magnethodydrodynamics (MHD), on a consistent basis. The objectives of this study are to provide 1) descriptions of the state of the art for these conversion systems and projections of future technological advances; and 2) an assessment of the performance of each of these power conversion systems using medium-Btu fuel gas. This document presents the results of Task 1 (Assessment of Systems Technology Status), Task 2 (Description of Advanced Power Systems), Task 3 (Evaluation of Advanced Power System Utilizing Coal-Derived Medium-Btu Gas), and Task 4 (Evaluation of Advanced Power Generation System Integrated with a Coal Gasifier Producing Medium-Btu Fuel Gas).

The work described in this report has been performed at United Technologies Research Center (UTRC) and Burns and Roe, Inc. (B&R) under Contract No. NAS8-33996 with George C. Marshall Space Flight Center of the National Aeronautics and Space Administration (MSFC).

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1.1 Assessment of Fuel Cell Power System Status

The objectives of this technology assessment have been to develop a data base reflecting the current status of both phosphoric acid and molten carbonate fuel cell power generation system technology and to provide an appraisal of probable future development of the technology. The scope of this assessment effort included: a literature review of the current status of fuel cell technology; summarization of fuel cell power generation system operational behavior and experience; and a projection of the future course of fuel cell power generation system development with an indication of potential challenges which must be met to fulfill the projections. The objectives of this technology assessment effort as stated above provide the format for the following commentary.

Prior to reviewing the commentary, however, it may be appropriate to first review the three appendices provided as background to the fuel cell assessment. Appendix A.1 includes a development of the electrochemical fundamentals associated with fuel cells and is provided for the purpose of introducing the key fuel cell performance relationships. Appendix B.1 contains a discussion of the potential advantages of a fuel cell power generation system to a utility. Appendix C.1 furnishes a review of the historical development of the fuel cell up to the present programs.

1.1.1 Current Technology Status and Operational Data Base of Fuel Cell Power Systems

A fuel cell power system consists of a fuel processor that produces H₂-rich fuel from raw and primary fuels, a power section that converts the chemical energy in the gaseous feeds to dc power, and a power conditioner that converts dc to ac power (Fig. 1.1.1-1). To utilize the heat rejected from the fuel cell power system, a cogeneration system to supply thermal energy or a bottoming power generation system to provide additional electrical output could be added. A hydrocarbon fuel (natural gas, mined coal, solid waste, refined petroleum products or resids, synthetic natural gas, naphtha, methanol, etc.) plus steam, and heat recycled from the fuel cell section are fed into the fuel processor to generate a gas rich in hydrogen. The fuel processor could be a reformer, a gasifier, or a hydroprocessor, depending on the hydrocarbon fuel used. The hydrogen-rich mixture is then fed to the anode side of the fuel cell power section which is formed by fuel cells connected electrically in series to generate the desired voltage. The dc power obtained is then processed by the power conditioner (at about 95-percent efficiency) to produce the ac power to meet the user requirements.

Fuel cells and fuel cell power systems are usually classified according to the type of electrolyte or ion conducting media used. In the current study effort two fuel cell types are of special interest: the phosphoric acid fuel cell and the molten carbonate fuel cell. The technology of the phosphoric acid fuel cell has been highly developed. As such it is regarded as a

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FUEL CELL POWER SYSTEM

first generation fuel cell. Because of the near-term application potential of the phosphoric acid fuel cell, phosphoric acid fuel cell power systems will be discussed first followed by a discussion of the technology of the more long-term, second generation molten carbonate fuel cell power systems.

The topic of the commentary in the following sections is the current status of both phosphoric acid and molten carbonate fuel cell power system. The format for the discussion of these fuel cell power systems will be similar; a brief overview of the principles of operation will be followed by a commentary relating the important parameters which effect performance and identifying the current state of technology which defines these parameters. The material presented has been collected from the recent open literature, a bibliography of which is provided in a later section. One work was especially useful, however, and should be cited and that is (The) Handbook of Fuel Cell Performance prepared under DOE Contract No. EC-77-03-1545 by T. G. Benjamin, E. M. Cawara, and L. G. Marianowski.

1.1.1.1 Phosphoric Acid Fuel Cell Power Systems

State-of-the-art phosphoric acid fuel cells operate at 150 to 200 C and 1 to 3 atmospheres. The electrical operating point is 0.55 to 0.65 volts/cell at 100 to 400 mA/cm² of electrode area. The present phosphoric acid cell is adequate for utility applications. The fuel cell power system efficiency is 32 to 40 percent which is consistent with reasonable stack costs provided the stacks are made by mass production techniques.

Although small phosphoric acid stacks have operated for extended periods, the desired 40,000 hours has yet to be demonstrated. A 4.8-MW demonstration plant containing all the equipment and facilities necessary to convert fuel into ac electrical necessary to convert fuel into ac electrical power is being constructed and will be operational this year.

Principles of Operation

In the phosphoric acid cell (see Fig. 1.1.1.1-1) gaseous hydrogen in the fuel diffuses through the porous anode to a reaction site at the electrode (solid)/electrolyte (liquid) interface where it is electrochemically oxidized according to:

$$H_2 + 2H^+ 2e^-$$
 (1.1.1.1-1)

The electrons are transported through the external circuit, and the hydrogen ions are conducted through the electrolyte to the cathode reaction sites. Oxygen, which has diffused through the cathode, reacts with the hydrogen ions and electrons (Eq. 1.1.1.1-2), and the product water diffuses back out of the cathode:

$$1/2 O_2 + 2H^+ + 2e^- + H_2O(g)$$
 (1.1.1.1-2)

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PHOSPHORIC ACID FUEL CELL

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The net reaction is:

$$H_2 + 1/2 O_2 + H_2O(g)$$
 (1.1.1.1-3)

Cell Components

A sketch of a conventional phosphoric acid fuel cell is displayed in Fig. 1.1.1.1-2. The cell is composed of a phosphoric acid electrolyte sandwiched between two porous electrodes. The electrolytes are retained by either a thin inert matrix or a thin pad acting as a blotter. The electrodes are porous carbon catalyzed with small amounts of platinum. They are backed with porous conductive support sheets. Sandwiching the entire cell are separator plates which as the name implies separate adjacent cells. The separator plates are ribbed to provide channels for the fuel and oxidant to flow to the anode and cathode, respectively. A large number of these individual cells are then assembled in series to provide the desired output power and voltage. For example, the 4.8-MW demonstrator to be used in Con Edison has fuel stacks with approximately 460 individual cells.

One of the goals of past and current fuel cell programs has been to reduce cell stack cost. One of the most notable achievements has been the reduction of platinum catalyst required in the electrodes. Current cells require less than 4 percent of the platinum used in earlier cells $(0.75 \text{ mg/cm}^2 \text{ as compared}$ to 20 mg/cm² of active cell area). This has been accomplished with minor effect on cell performance (Fig. 1.1.1.1-3).

Another cell structure advance which promises to reduce fabrication costs is the integral ribbed substrate (Fig. 1.1.1.1-4). This cell concept will also provide increased electrolyte capacity. In part, the cost reduction is brought about by the substitution of a 2-3 day process in place of the conventional 2-week molding and curing process which produces a complex bipolar/separator plate. Aside from being complex, the molded processing method places a limitation on the size of the plate that can be fabricated. Eliminating the size constraint on the cell structure thereby provides the possibility to go to a larger and more cost effective cell.

The unique features of the integral ribbed substrate concept are: 1) replacing with a simple impervious flat plate the complex bipolar/separator (which normally contains a ribbed reactant flow field on either side perpendicular to each other); and 2) transferring the function of each solid ribbed flow field onto a ribbed porous element which, after deposition of catalyst on one side, behaves like an electrode and which is also designed to act as an electrolyte reservoir.

General Performance Considerations

The magnitudes of the anode, the cathode (activation plus concentration), and ohmic polarizations for a typical phosphoric acid fuel cell are shown in



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PHOSPHORIC ACID FUEL CELL POLARIZATION CURVES

a) 160°C, 1 Atm, AIR, PLATINUM LOADING OF 20 MG/CM², H2FUEL

b) 190°C, 3 Atm,AIR, PLATINUM LOADING OF 0.75 MG/CM2, H2 FUEL

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INTEGRAL RIBBED SUBSTRATE CELL CHARACTERISTICS

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Fig. 1.1.1.1-5. The upper curve represents the cell performance that could be obtained if ohmic (approximately 15 mV/100 mA/cm²) and anode (approximately 5 mV/100 mA/cm²) polarizations were eliminated. It also demonstrates the significant gains possible if the cathode polarization (approximately 400 mV/300 mA/cm²) could be reduced. The relative magnitudes of the anode and cathode polarizations illustrate why most recent cell electrochemistry research is concentrated on the cathode.

Although the phosphoric acid fuel cell is much more developed than any other fuel cell system, it is difficult to find clear parametric performance data in the open literature. Therefore, in the following discussion on the effects of various performance parameters on fuel cell performance, a baseline system is used to illustrate the effect of parameter variations. The baseline cell operates with H_2 and O_2 at low utilization, atmospheric pressure, and 175 C. Its performance is shown Fig. 1.1.1.1-6.

Temperature Effects

Increasing temperature will enhance mass transfer, increase the reaction rate, and usually decrease cell resistance, thus reducing polarization and increasing cell performance. In the case of the acid cell, the additional benefit of increased CO tolerance at higher temperature is realized. On the other hand, increasing temperature results in increased corrosion, catalyst sintering and recrystallization, and electrolyte loss due to evaporation, so a compromise is required. Current cells operate at 190 C but work is in progress to increase this temperature level.

The increase in cell performance which could be realized by increasing operating temperature is shown in Fig. 1.1.1.1-7 for a cell burning either reformed methane (RM-1) with air, H_2 with air, or H_2 with O_2 all at 50 psia and 300 amp/ft² current density. It can be concluded from this illustration that modest temperature increases on the order of 30 C above the current level of 190 C could produce significant (up to 10 percent) performance increases.

Pressure Effects

There are multiple incentives for operating a fuel cell at elevated pressure, including integration with fuel processors, increased reactant partial pressure (Nernst gains), and higher gas solubilities in the electrolyte. There are also problems associated with high-pressure operation such as increased component corrosion resulting from the higher voltage. Phosphoric acid fuel cells operate in the range of 1 to 10 atmospheres.

Increasing cell pressure will result in higher cell performance because of decreased activation polarization at the cathode due to increased oxygen and water partial pressure. The water dilutes the acid resulting in a higher exchange current density (see Appendix A.1). An additional gain is realized in the open circuit potential due to the increased partial pressure of water and oxygen.

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PHOSPHORIC ACID FUEL CELL POLARIZATION LOSSES

190° C, 3 Atm, AIR, PLATINUM LOADING, 0.75 MG/CM², H₂ FUEL

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BASELINE PHOSPHORIC ACID FUEL CELL PERFORMANCE 1 Atm, 175°C, H₂ FUEL, O₂ OXIDANT, LOW UTILIZATION

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EFFECT OF TEMPERATURE ON PHOSPHORIC ACID FUEL CELL PERFORMANCE 50 PSIA, 300 ASF, 190° F ANODE DEWPOINT

The increase in cell performance resulting from increasing pressure is shown in Fig. 1.1.1.1-8. A correlation describing the relationship between cell voltage change, ΔV , and pressure ratio, P_2/P_1 , as shown in Fig. 1.1.1.1-8, is:

$$\Delta V_{p} = 142 \log (P_{2}/P_{1}) \qquad (1.1.1.1-4)$$

This correlation appears to be valid at least over the range from 190 C to 220 C because the effect of pressure on cell performance is independent of temperature over this range.

Reactant Utilization Effects

High reactant utilization in a fuel cell results in increased concentration polarization, which reduces efficiency. Decreasing reactant utilization or inlet concentration, however, results in decreased cell performance because of increased concentration polarization (diffusion) and Nernst losses. To achieve high system efficienies the reactant must be utilized, either in the fuel cell or elsewhere in the system. For example, in a complete fuel cell system, anode exhaust containing unreacted hydrogen is often burned to supply heat to the fuel processor. Present utilizations are on the order of 70 percent oxidant and 85 percent fuel.

Oxidant

At a constant cathode potential the current varies linearly with oxygen partial pressure. Data on the gains associated with using oxygen at the cathode instead of air are contained in Figs. 1.1.1.1-7 and 1.1.1.1-8. The data in these figures are represented by:

$$\Delta V_{0_2} = 103 \log \frac{(\bar{P}_{0_2})_2}{(\bar{P}_{0_2})_1}$$
(1.1.1.1-5)

where $(\bar{P}_{0_2})_1$ and $(\bar{P}_{0_2})_2$ are average oxygen partial pressures at the cathode.

Fuel

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Because the phosphoric acid fuel cell anode reaction is essentially reversible, the effects of fuel composition and utilization are less critical to cell performance. The available data concerning the effects of fuel composition are more difficult to interpret because of the poisoning effect of carbon monoxide, which is contained in virtually all gaseous fuel cell feeds derived from hydrocarbon (liquid or coal) feedstocks. The cell performance change with respect to hydrogen partial pressure is derived from the data in Fig. 1.1.1.1-9. ŝ

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EFFECT OF PRESSURE ON PHOSPHORIC ACID FUEL CELL VOLTAGE 375°F CELL TEMPERATURE , 190°F ANODE DEWPOINT, 300 ASF

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EFFECT OF REACTANT UTILIZATION ON PHOSPHORIC ACID FUEL CELL PERFORMANCE

$$\Delta V_{\rm H_2} \ (mV) = 77 \ \log \frac{(\bar{P}_{\rm H_2})_2}{(\bar{P}_{\rm H_2})_1} \ (1.1.1.1-6)$$

The hydrgoen partial pressures are average values at the anode.

Reactant Utilization

The effect of reactant utilization on a fuel cell is shown in Fig. 1.1.1.1-9. There is a definite increase in cell voltage resulting from high oxidant utilization while the level of fuel utilization has little effect on fuel cell voltage. This data resulted from a test to evaluate a possible approach to achieve high utilization without incurring increased concentration polarization.

Reactant Composition Effects

The ever widening selection of hydrocarbon fuel processing techniques have presented the fuel cell with a wide range of fuels from which hydrogen can be obtained. Knowledge of the various compounds especially the carbon, sulfur, and nitrogen compounds which could effect the fuel cells is important.

Carbon Compounds

Anode feeds derived from hydrocarbons will contain varying amounts of methane, carbon dioxide, and carbon monoxide depending upon the fuel processing techniques used. Both methane and carbon dioxide are inert in the phosphoric acid fuel cell acting as a diluent. Carbon monoxide, however, will poison the platinum catalyst. Consequently, the level of CO in the fuel gas must be controlled. It is possible to mitigate somewhat the effect of CO on the anode by increasing the cell temperature (Fig. 1.1.1.1-10). Although relatively minor temperature increases will extend the acceptable CO level significantly, the reverse is also true, a minor temperature decrease in an operating cell could quickly lead to anode poisoning.

The low CO levels necessary for the operation of the acid fuel cell can be achieved by conventional water-gas shift technology. Utilizing two beds operated at 315-485 C and 175-350 C, respectively, the CO levels can be reduced to 0.05-0.5 percent. With recent developments in sulfur-tolerant shift catalysts, these low concentrations can also be achieved in the presence of H₂S.

Sulfur Compounds

Advanced fuel processors operating on sulfur-containing fuels will produce H_2S that will enter the anode if not first removed. Testing of a fuel cell running on fuel gas containing 130 to 200 ppm H_2S was characterized by long

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FIG. 1.1.1.1-10



EFFECT OF TEMPERATURE ON PHOSPHORIC ACID FUEL CELL ANODE CO TOLERANCE

periods of stable operation followed by rapid decay. The original performance could be restored by removing either CO or H_2S from the anode gas, indicating a synergistic effect, or by increasing the cell temperature from 375 to 400 F. The mechanism of H_2S -CO poisoning is not well understood. Post-test analysis of electrodes from shorter term cells showed no observable changes attributable to exposure to H_2S containing anode gas. Work to broaden the understanding of the CO- H_2S interaction in the phosphoric acid fuel cell is continuing.

While the technology for H_2S removal is commercially available, the necessity to remove large amounts of H_2S may cause considerable logistic problems or unrealistic system complexity, particularly with small dispersed power stations.

Nitrogen Compounds

Nitrogen will be present at the cathode of an air-consuming fuel cell, and nitrogen-containing hydrocarbon feedstocks will introduce nitrogen compounds (NH_3, HCN, NO_x) into the anode. In addition, nitrogen compounds can result if nitrogen-free hydrocarbons are processed with air such as in autothermal reforming or coal gasification. The nitrogen in this case will be primarily molecular nitrogen, which acts as a diluent. Data on the interaction of nitrogen compounds other than ammonia have not been found in the literature.

Ammonia reacts with the phosphoric acid electrolyte to form ammonium dihydrogen phosphate (ADHP) according to:

$$H_3PO_4 + NH_3 + (NH_4) H_2PO_4$$
 (1.1.1.1-7)

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The formation of the ADHP results in a performance loss, possibly by changing the conductivity of the acid or by reducing the activity of the platinum by covering the active sites.

Summary

Data for other possible impurities have not been reported but estimates of maximum allowable levels are summarized in Table 1.1.1.1-1.

Fuel Processor and Power Conditioner

The fuel processor and the power conditioner are the two other subsystems beside the fuel cell power subsystem which comprise a fuel cell power generation system.

Fuel Processor

A number of studies have indicated fuel cell systems could operate efficiently with the potential coal-derived fuels. The principal effect of utilizing different products will be on the amount and complexity of fuel processing equipment required. This is illustrated in Table 1.1.1.1-2.

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TABLE 1.1.1.1-1

ESTIMATED MAXIMUM ALLOWABLE IMPURITY LEVELS PHOSPHORIC ACID FUEL CELLS

Impurity	<u>Maximum Limit</u>
co2	diluent
сн ₄	diluent
N ₂	diluent
со	Figure 1.1.1.1-10
H ₂ s, cos	100 ppm
c ₂ +	100 ppm
c ₁ -	< 1 ppm
NH ₃	< 1 ppm
Metal ions (Fe, Cu, etc.)	nil

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TABLE 1.1.1.1-2

FUEL PROCESSING EQUIPMENT REQUIREMENTS FOR DIFFERENT COAL PRODUCTS IN PHOSPHORIC ACID POWERPLANTS

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Fuel Processing Equipment

Coal Product	Reformer	Shift Converters	Water Recoverv	Water Conditioning	Steam Generation
High Btu Gas (SNG) 1000 <u>BTU</u> Ft ²	X	X	X	X	X
Medium Btu gas (H ₂ ; CO) 325 <u>BTU</u> FT ³	-	X	X	X	X
Medium Btu Gas (H ₂) 325 <u>BTU</u> Ft ³		-	-	-	-
Liquid Product (Naphtha or Methyl Fuel)		x	x	X	x

Alternatives to the steam reforming are being investigated to achieve the capability of processing fuels containing substantial levels of sulfur and unsaturated hydrocarbons. Emphasis is being placed on operation at higher temperature to achieve near complete fuel conversion over suitable catalysts.

Power Conditioner

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The power conditioning subsystem converts the dc current from the fuel cell stacks to alternating current and controls its flow into the utility network. This function is accomplished through the use of solid state inverters. The ac voltage output from the inverters is stepped up through transformers to the desired level for utility distribution. Elimination of the power conditioning equipment will result in a dc voltage output which varies from a maximum value at low loads to approximately 75 percent of this voltage at rated power.

The power conditioner now available appears to be satisfactory for current fuel cells and more advanced cells.

1.1.1.2 Molten Carbonate Fuel Cell Power Systems

State-of-the-art molten carbonate cells operate at 650 to 700 C and up to 10 atmospheres. The cell voltage is 0.7 to 0.9 volts/cell at 100 to 150 amps/ ft^2 . The present molten carbonate fuel cell technology is at least five years behind the phosphoric acid fuel cell technology. A fuel cell power system utilizing a molten carbonate fuel cell could be 45 to 50 percent efficient. It is too early in the development of the molten carbonate fuel cell to accurately predict the final cost for the power system.

The endurance of molten carbonate fuel cells is still under investigation with small-scale cells logging more than 10,000 hours. To date the benchmarks in molten carbonate development have included thermal cycling of a 20-cell stack with 12" x 12" cells and 1000 hours of operation of a 10" x 10" cell.

Principles of Operation

In the molten carbonate cell (see Fig. 1.1.1.2-1) the electrolyte is a mixture of alkali metal carbonates (a molten ionic conductor at the cell operating temperatures of 600 to 700 C) and ceramic particulates that retain the liquid. Fuel enters the anode, and the H_2 reacts at the anode/electrolyte interface according to:

$$H_2 + CO_3 \longrightarrow H_2O + CO_2 + 2e^-$$
 (1.1.1.2-1)

* Carbon monoxide in the fuel can also be oxidized, but a more favorable reaction for the conversion of CO is via conversion to hydrogen by the shift reaction, $CO + H_2O \rightarrow CO_2 + H_2$. The hydrogen formed is then consumed according to 1.1.1.2-1. .

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$$H_2 + CO_3^{*} = H_2O + CO_2 + 2e^{-1}$$

 $CO_2 + 1/2 O_2 + 2e^{-1} = CO_3^{*}$
 $H_2 + 1/2 O_2 = H_2O$



MOLTEN CARBONATE FUEL CELL

The product H_20 and CO_2 diffuse out of the electrode. At the cathode, the following reaction takes place:

$$CO_2 + 1/2 O_2 + 2e^- + CO_3^-$$
 (1.1.1.2-2)

Note that CO_2 is formed at the anode and consumed at the cathode. For an efficient molten carbonate fuel cell system, the CO_2 in the anode effluent must be recycled back to the cathode. This could be accomplished by catalytic combustion of the anode effluent to CO_2 and H_2O plus N_2 in the combustion air), followed by mixing with air to produce the cathode feed.

Cell Components

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A schematic of one repeating element in a stack of molten carbonate fuel cells is shown in Fig. 1.1.1.2-2. The electrolyte tile is held in place between the two stainless-steel cell housings that form the anode and cathode cavities. The holding force maintains low-resistance contacts between the electrodes and the stainless steel current collectors and causes the housings to bite into the electrolyte tile, forming a wet-seal that prevents leakage of fuel from the anode chamber and oxidant from the cathode chamber. In a stack of several cells (series connection) the cell separator must be able to withstand both the high-humidity reducing environment of the anode and the oxidizing environment of the cathode.

One of the major goals of current molten carbonate programs is to fabricate cell stacks containing a large number of cells of sufficient size to produce a significant power level. One of the most notable achievements in molten carbonate fuel cell development has been the extension of current density by a factor of four yet maintaining a promising level of cell endurance. Other improvements are: 1) electrolyte change to the current Li_2CO_3/K_2CO_3 alkali carbonate eutectic which increased initial cell performance; 2) the anode structure was stablized by adding sintering inhibitor to the nickel, and 3) the use of thinner electrolyte tiles reducing iR losses.

FIG. 1.1.1.2.2

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MOLTEN CARBONATE FUEL CELL CROSS SECTION

General Performance Considerations

For a cell operating on reformed natural gas fuel at 75-percent total H_2 utilization and on an oxidant mixture of 70-percent air and 30-percent CO_2 utilization, at 200 mA/cm², losses are approximately 68 mV ohmic and 260 mV polarization with anode polarization losses being greater than those at the cathode. Of course, these losses depend on the fuel cell operating variables. Consequently, for the purposes of this discussion, molten carbonate fuel cell performance at atmospheric pressure and 650 C, using a cell operating on simulated reformed methane fuel (the composition is 60 mol X H₂ 10 mol X CO, 7.4 mol X CO₂, and 22.6 mol X H₂O) at low utilization and on an oxidant consisting of 67-mole percent CO_2 and 33-mole percent O_2 at low utilization, will be considered as baseline. The electrolyte tile composition is 40-weight percent inert LiAlO₂ particles to provide structural integrity and 60-weight percent binery eutectic mixture of Li₂CO₃ and K₂CO₃ as the electrolyte. The baseline performance is shown in Fig. 1.1.1.2-3.

Temperature Effects

As the molten carbonate fuel cell operating temperature is increased through its range of 600 to 700 C, hydrogen is consumed for a gas in water gas shift equilibrium. Because standard cell potential also decreases with increasing temperature, the open-circuit potential decreases as temperature increases: However, ohmic, activation, and concentration polarizations all decrease, more than offsetting the decrease in the open-circuit potential. The net result is an increase in cell performance with the anode being only slightly affected by changes in cell temperature.

A plot illustrating the potential voltage gains is provided in Fig. 1.1.1.2-4. There is a discontinuity in the curve at 650 C with a diminishing gain in performance at temperatures greater than 650 C. This diminishing gain coupled with increased electrolyte loss resulting from evaporation and increased component corrosion, point to an optimum operating temperature of 650 C for molten carbonate fuel cells with an electrolyte tile composed of 40-percent LiAlO₂, 28-percent K₂CO₃, and 32-percent LiCO₃ by weight.

Pressure Effects

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Cell performance increases with pressure because of increased reactant partial pressures, increased reactant solubilities, and increased mass transport rates. Higher pressures tend to favor carbon formation and consumption of CO and H_2 fuel. The addition of water controls carbon formation, because it reduces the amount of CO and methanation. However, strict pressure control is required to prevent reactant leakage through the electrolyte and seals, to provide compatibility with the fuel processor, and to inhibit the possible corrosion of cell components at the higher pressure which molten carbonate fuel cells are expected to operate (1 and 10 atmospheres).

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BASELINE MOLTEN CARBONATE FUEL CELL PERFORMANCE

1 Atm,650 ° C,67 Moi % CO 2/33Moi % O2 OXIDANT AT 15% UTILIZATION, AT 25% UTILIZATION

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EFFECT OF TEMPERATURE ON MOLTEN CARBONATE FUEL CELL PERFORMANCE

1 Atm, 200 mA/CM², 70 Mol % AIR/30 Mol % CO₂ OXIDANT AT 50 % UTILIZATION ,75 % FUEL UTILIZATION

The effect of cell pressure on cell performance is displayed in Fig. 1.1.1.2-5. For the product gas of an air-blown, low-Btu coal gasifier, the relationship between changes in cell performance and operating pressure in atmospheres is:

$$\Delta V_{\rm p}({\rm mV}) = 70 \log \frac{P_2}{P_1} \qquad (1.1.1.2-3)$$

and for reformed methane, it is

$$4V_{\rm p}({\rm mV}) = 83 \log \frac{{\rm P}_2}{{\rm P}_1}$$
(1.1.1.2-4)

Because these two cases represent a wide range of gas composition, other fuels should obey the average relationship:

$$\Delta V_{\rm p}({\rm mV}) = 76.5 \log \frac{P_2}{P_1} \qquad (1.1.1.2-5)$$

Reactant Utilization Effects

The effect of reactant utilization on molten carbonate fuel cell performance is difficult to analyze because of the participation of CO at the anode via the water-gas shift reaction and because of consumption of CO_2 and O_2 at the cathode.

Oxidant

According to the Nernst equation, (see Appendix A.1) performance should be related to oxidant $(CO_2 \text{ and } O_2)$ partial pressure. However, at low oxidant partial pressure (high utilization), the gains are caused by diminishing mass transfer effects in addition to the Nernst gains. As the partial pressure increases, mass transfer effects become negligible and the gains become governed by Nernst effects.

Fuel

The performance related to the fuel partial pressure is directly related to the Nernst equation.

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AT 50 % UTILIZATION,75 % FUEL UTILIZATION

Reactant Utilization

The data of Fig. 1.1.1.2-6 is representative of data published on reactant utilization. The data indicate that increasing oxidant utilization (decreasing oxidant partial pressure) causes less cell potential loss than increases in fuel utilization. Once again a compromise is required, because operating at low fuel utilization increases cell performance but decreases the current efficiency. Current molten carbonate fuel cells commonly run at 75 to 85-percent fuel utilization and 50-percent oxidant utilization.

Reactant Composition Effects

The question of molten carbonate fuel cell tolerance to coal-derived contaminants is inherently complex. One problem is that the identity and concentration of expected contaminants is highly variable and depends upon the particular coal sample and its pretreatment. Further, even when well defined "synthetic" contaminants are to be evaluated -- the effect on the fuel cell may be dependent upon the concentrations of other components in the fuel or oxidant, the degree of fuel/oxidant utilization, the cell voltage, and the cell current density. Finally, experiments with low contaminant concentrations are difficult to carry out and evaluate conclusively.

Many discussions have centered on integrating the molten carbonate fuel cell with advanced fuel processors because of its tolerance to CO at the anode and CO₂ at the cathode. Other compounds likely to be present in the effluent of advanced fuel processors include NH₃, NO_x, SO_x, H₂S and COS.

Carbon Compounds

Solid carbon can form in the cell or in system plumbing. The addition of water prevens carbon deposition by reducing CO concentrations through the water-gas shift reaction. Carbon dioxide present in the anode feed and produced at the anode acts only as a diluent, reducing the hydrogen partial pressure. In fact, a small amount of CO_2 is required to maintain the LiCO₃/K₂CO₃ electrolyte composition. Of course, CO_2 is required at the cathode. Although carbon monoxide dilutes the hydrogen, it is also consumed indirectly through the water-gas shift reaction.

Methane at the anode can be reformed to hydrogen by adding sufficient water to the fuel gas. If the methane is not reformed, fuel cell efficiency will decrease; energy in the methane must be recovered elsewhere in the system to maintain the total system efficiency.

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EFFECT OF REACTANT UTILIZATION ON MOLTEN CARBONATE FUEL CELL PERFORMANCE

1 Atm, 650°C, 160 mA/CM², 70 Mol % AIR/30 Mol % CO₂ OXIDANT, SIMULATED REFORMED METHANE FUEL

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Sulfur Compounds

Sulfur initially reaches the cathode in the form of SO_2 and SO_3 . Sulfur entering the cathode will tend to form SO_4^{-} ions in the electrolyte. These ions will be transported through the electrolyte along with the predominant CO_3^{-} ions, reacting at the anode to form additional H₂S and COS. Thus a concentration of sulfur will take place in the cell, reaching some steady state distribution along the cell with the highest sulfur levels at the anode outlet and cathode inlet.

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The presence of sulfur in the cell creates several caverns. A first concern would be what effects the sulfur will have on the anode where the sulfur, as predominately H_2S , can sulfidize the nickel electrode, poison the surface of the catalyst by chemisorption, or create a mixed potential by entering into an electron exchange reaction. A second concern is the effect of sulfur on the cathode compartment, where as SO_2 or SO_3 it might cause sulfidation of the electrode. A third concern is the effect of sulfur in the electrolyte where it can affect the transport of the CO_3^{-1} ions, alter the chemistry of the LiAlO₂ support material within the electrolyte matrix (tile), and form sulfides, sulfites and sulfates. A final concern is the effect of sulfur on corrosion of cell hardware and system plumbing.

Nitrogen Compounds

Molecular nitrogen at the anode or cathode acts as diluent. The presence of NH₃ and NO_x is also expected at both anode and cathode, but no data on the effect of these impurities on cell performance are available, although the $K_2CO_3/Li_2CO_3/Na_2CO_3$ eutectic is known to absorb NO.

Chlorine Compounds

Like sulfur, chlorine compounds are anticipated in a coal-derived fuel gas. Recent short-term data indicate that anode performance is only slightly affected after 400 hours on a fuel gas containing 100 ppm HCl, but cathode performance is catastrophically affected by 100 ppm HCl in the oxidant gas. Stable cathode performance with 10 ppm HCl has been demonstrated for 120 hours. Although longterm trends have not been established, it appears that the decay results from both hardware corrosion and chlorination of the electrolyte.

Summary

A summary of the estimates for maximum allowable contamination limits is displayed in Table 1.1.1.2-1.

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TABLE 1.1.1.2-1

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ESTIMATED MAXIMUM ALLOWABLE IMPURITY LEVELS MOLTEN CARBONATE FUEL CELLS

Impurity	<u>Maximum Limit</u>
co ₂	diluent
CH ₄	diluent/fuel
co	fuel
N ₂	diluent
H ₂ s, so ₂ , cos	< 1 ppm
c ₂ +	unknown
HCl (anode)	100 ppm
HCl (cathode)	10 ppm
NH ₃ m NO _x	unknown
Metal ions (Fe, Cu, etc.)	unknown

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1.1.2 Future Fuel Cell Development

The course of future fuel cell development has already been charted. Its milestones will be the successful achievement of the goals and the solving of the problems identified in current fuel cell programs. The following section will provide an identification of some of the major fuel cell investigations and an indication of the program objectives associated with the development of phosphoric acid or molten carbonate fuel cells. In addition a scenario of projected fuel cell development will be furnished. The sources for the data provided herein are presented in the accompanying Bibliography (Section 1.1.3).

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1.1.2.1 Fuel Cell Programs

The major participants in the fuel cell program for electric utilities are EPRI (Electric Power Research Institute), DOE (Department of Energy), ANL (Argonne National Laboratory), TVA (Tennessee Valley Authority), UTC (United Technologies Corporation), GE (General Electric), IGT (Institute of Gas Technology), Westinghouse, and ERC (Energy Research Corporation).

EPRI Program

EPRI as the electric utility centralized research coordinating organization quite legitimately is playing a major role in fuel cell development. Not only does EPRI sponsor fuel cell programs but also assists in coordinating and guiding utility fuel cell research and development.

DOE/ANL Program

ANL has been assigned as technical manager of the DOE advanced fuel cell program, especially those utilizing carbonate as the electrolyte.

TVA Program

TVA is engaged in evaluating the feasibility of using coal-derived fuel gas as a fuel cell fuel. This evaluation will be carried out in part by actually operating a phosphoric acid fuel cell on coal-derived fuel gas.

UTC Program

UTC is working toward commercializing phosphoric acid fuel cell power plants. This activity includes, among others, participating with EPRI and DOE in the 4.8-MW demonstrator in New York City (see Appendix C.1), evaluation of manufacturing techniques to reduce phosphoric acid fuel cell fabrication costs, investigation of phosphoric acid technology to improve performance and increase lifetime, and development of molten carbonate fuel cells.

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IGT Program

IGT is investigating the molten carbonate cell and its utilization with gasified coal.

GE Program

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GE is active in molten carbonate cell investigation. Of special interest is the technical and economic problems of developing fuel assemblies and stacks. In conjunction with IGT, GE is evaluating cell lifetime.

Westinghouse Program

Westinghouse is currently involved in the investigation of a unique fuel cell cooling concept called DIGAS. This approach to stack cooling has the p-tential of improving performance while at the same time increasing endurance.

ERC Program

ERC has been active in phosphoric acid fuel cell development. They have teamed with Westinghouse and DOE to evaluate a 120-kw on-site fuel cell which will provide for electric power and thermal energy.

1.1.2.2 Program Objectives

Because of the different state of development of the phosphoric acid and molten carbonate fuel cells, the program goals differ and will be treated separately.

Phosphoric Acid

Although the phosphoric acid fuel cell system is on the verge of large-scale demonstration, a number of areas for possible improvement are still receiving attention. For example:

- The platinum catalyst is expensive and suffers a reduction in surface area and performance due to sintering. The CO-poisoning problem mentioned earlier has been somewhat mitigated by operation at higher temperatures, however, increasing the temperature also increases the sintering. CO-tolerant catalysts such as mixtures of Pt, Rh and WO₃ have also been used.
- 2. Electrolyte is lost because of evaporation. This problem can be alleviated by incorporation of an electrolyte reservoir in the cell. However, this reservoir can increase concentration polarization in the cell, and the evaporated acid condenses dcwnstream of the cell, creating corrosion problems.

3. The most difficult challenge concerns the irreversibility (severe activation polarization) and lack of understanding of the reduction of oxygen at the cathode. This area also offers considerable opportunity for improvement of the phosphoric acid fuel cell performance.

In addition to continued effort in the above areas it also necessary to continue development of the fuel cell system so as to meet the following cost and endurance goals for a commercially acceptable power plant:

1. Implementation of an improved cell stack concept which is necessary to reduce cell stack manufacturing costs and to provide sufficient electrolyte capacity for 40,000 hours of operation. -

- 2. Incorporation of more corrosion resistant and stable materials and supported electrocatalysts in the fuel cell which are necessary to meet performance and life goals (9300 Btu/kWh heat rate for 40,000 hours).
- 3. Increasing cell operation pressure (to approximately 100 psia) and temperature (by approximately 30°F) to substantially improve power plant performance and reduce cost.

Accomplishment of these goals will produce an evolutionary improvement in technology. An example of such an improvement is shown in Fig. 1.1.2.2-1. The effect of a moderate increase in temperature and an increase in pressure has produced an improvement in performance as compared to the 4.8-MW demonstrator plant.

As a result of this evolutionary advance of technology it is anticipated that a phosphoric acid fuel cell, a decade hence, will not be unlike the current fuel cell.

Moiten Carbonate

Although molten carbonate fuel cell technology has made impressive advances in the last decade, a number of questions concerning cell performance and life remain:

- 1. The sulfur and chlorine tolerance of the cell is only a few parts per million; the tolerance to other impurities has not been defined.
- 2. Electrolyte losses result in decreases in electrolyte tile strength and bubble pressure, increased tile porosity, and diffusion of reactants across the tile causing decreases in cell voltage and

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FIG. 1,1.2.2-1



PERFORMANCE OF CELL AT VARIOUS OPERATING CONDITIONS

efficiency. A partial solution to the evaporation problem is the incorporation of a reservoir with enough electrolyte to achieve the design life of 40,000 hours. The electrolyte may not be evenly distributed, however, and redistribution can cause performance loss. In addition, new tile and electrode structures and materials that inhibit carbonate evaporation are being sought.

- 3. Electrolyte creepage and corrosion result in failure of the wet-seal and leakage of reactants out of the cell. The wet-seal is now protected by aluminizing although it is uncertain whether this approach will be effective for 40,000 hours.
- 4. Effects of structural changes in the electrolyte support (LiAl0₂) on cell performance have not been defined.
- 5. Cost-effective cell components (tile, electrodes, and hardware) and manufacturing techniques must be identified.
- 6. Transfer of the CO_2 from the anode to the cathode is desirable with physical separation of the CO_2 from the rest of the anode exhaust preferred. To date studies on the approach to achieving this are incomplete.

Programs are underway addressing these development areas. DOE in its program anticipates development of the molten carbonate fuel cell power plants to a point where 1) operation of a prototype fuel cell stack with simulated coal-derived fuel gas is possible by 1982; 2) operation of a prototype stack occurs in 1983; and 3) operation of an integrated multi-MW test facility is accomplished by mid-1985.

Other goals for the molten carbonate fuel cell include: 1) a power density of 135 mW/cm² of active cell area using low-Btu coal-derived gas at a cell voltage of 0.85 volts and 75-percent utilization; 2) an operating time between major overhauls of 40,000 hours; 3) removal of the contaminant H_2S down to the 1-10 ppm level; and 4) development of practical cell stack fabrication techniques.

Attainment of these goals will continue the impressive advance of molten carbonate fuel cell technology during the last decade which has seen almost a quadrupling of current density (Fig. 1.1.2.2-2) with potentially long lifetimes.

1.1.2.3 Fuel Cell Development Scenario

One possible scenario of future fuel cell development has been advanced by EPRI (Fig. 1.1.2.3-1). This timetable takes into account the current work being done on both the phosphoric acid and molten carbonate cell. Several assumptions were made in developing the timetable. One of the most important is that it is

FIG. 1.1.2.2-2



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MOLTEN CARBONATE CELL PERFORMANCE HISTORY

DEVELOPMENT OF MATERIALS AND CELL CONFIGURATION COMMERCIAL FEASIBILITY (LIMITED PRODUCTION) COMMERCIAL FEASIBILITY (LIMITED PRODUCTION) FUTURE TECHNOLOGY (HEAT RATE: 7500 Blu/kWh) DEVELOPMENT OF IMPROVED HEAT RATE COMMERCIAL PROTOTYPE DEFINITION POWER PLANT DEMONSTRATION **4 5-MW DEMONSTRATION MODULE ADVANCED PHOSPHORIC CELL** CONFIRMATION IN STACK **CONFIRMATION IN STACK** MOLTEN CARBONATE CELL IMPROVED TECHNOLOGY 1-MW PILOT UNIT

EPRI TIMETABLE OF FUEL CELL DEVELOPMENT



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assumed that funding from the Federal government is forthcoming for the commercial prototype definition and commercial feasibility phases. Another equally important assumption is that a market will exit for the fuel cell. These two assumptions deserve further elaboration.

The first fuel cell power plants off the assembly line will not meet the goal of \$350/kW, they can be offered at that price only after some 500-1500 MW worth have been mass-produced in a factory. More likely the first unit will cost about \$1500/kW, with the cost of subsequent units decreasing by about 15% for each doubling of quantity, following a typical product learning curve. The eighth unit, for instance, should therefore cost \$920/kW. However, the worth of these units to an electric utility will be somewhere between \$330 and \$600/kW, depending on such factors as how severely a utility's siting options are restricted.

Even in the best of times, it is unlikely a utility would invest \$1500/kW for the first unit of an as yet immature technology. Nor are manufacturers likely to underwrite the 1500-MW learning curve. Consequently, in the EPRI scenario it is necessary for the federal government to assist the private sector in the commercialization process. Failure to do so could delay the timetable milestones to an unspecified later date.

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2.1 Description of Advanced Fuel Cell Power Systems

The two primary types of fuel cell discussed in Task 1 are the phosphoric acid and molten carbonate cells. Their relative performance and development status is shown in Fig. 2.1-1 in which the overall thermal efficiency of a number of integrated coal gasifier/fuel cell power plants are plotted as a function of the application time frame. As shown, it is possible to achieve efficiencies greater than 40 percent for systems integrating a coal gasifier with near-term phosphoric acid electrolyte fuel cells. In these, a gas turbine bottoming cycle was incorporated to recover power from the fuel cell exhaust. Systems studies with molten carbonate cells indicated potential efficiencies in the range of 45 to 60 percent depending on the degree of system integration, the type of bottoming cycle selected for waste heat recovery, and the technology assumed for the coal gasifier and the fuel cell.

The higher power plant efficiencies of molten carbonate systems result because at reasonable power densities, these cells offer higher efficiency than the nearterm phosphoric acid cell. Higher cell performance results from reduced activation polarization at high temperature. In addition, since molten carbonate cells operate at high temperature, they derive increased benefit from integration with the bottoming cycle.

In this section, the characteristics of each type will be reviewed briefly, the potential system configurations and optimum operating conditions reviewed, and the resulting candidate systems described.

2.1.1 Operational Characteristics

Probably the most significant characteristic of a fuel cell is the potential that can be developed across the cell and the reduction in that potential with increasing current density. For good efficiency a high cell potential is desired, while for minimum capital cost high current density, hence reduced potential is needed. Other characteristics that are important to system performance and economics are such things as fuel utilization, air utilization, temperature gradient and pressure drop of the cooling gas. Because all of these require optimization on a system level, they are discussed fully in Section 2.1.2 and only the basic cell characteristics are presented here.

2.1.1.1 Phosphoric Acid Cell Characteristics

The acid cell is typified by the 4.8-MW demonstration plant which is intended to expedite the introduction and acceptance of fuel cells for use as dispersed generators. This 4.8-MW unit is currently being installed at a Consolidated Edison Co. site in New York City. While it is generally believed that the performance of that system is typical of future acid cell units, advanced development work is aimed at reducing the manufacturing cost and improving the durability of the cells to make them competitive with other generation methods. That work is being sponsored by Niagra Mohawk Northeast Utilities, TVA, EPRI and DOE (Ref. 2.1-1).

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EVOLUTION OF INTEGRATED COAL GASIFIER/FUEL CELL POWER PLANTS

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For the acid cell, Fig. 2.1.1.1-1 compares the development cell voltage vs current density for two different operating conditions. It also shows the initial projected performance which correlates closely with the test data.

The cell performance decay characteristics were evaluated at both the estimated rated-power level and the estimated minimum-power level of a power plant. The rated-power level was 232 ASF, 93 psia, 405°F with a cell potential of 0.65 V to 0.70 V. The minimum-power level was at 93 psia, 405°F and a current density that resulted in 0.79 volts per cell average. This minimum-power level condition, which is an overstress condition on the components' electrochemical stability, was found to cause a greater decay rate than when operating at rated-power conditions. The performance diagnostic testing indicates that this greater decay rate is due to both a loss in cathode catalyst activity and an increase in cathode diffusion losses. Figure 2.1.1.1-2 shows the cell performance history and the decay rates experienced in a development stack. Advanced cathode catalyst development activities in small cell tests have identified candidate cathode catalysts which demonstrate lower performance decay rates than the GSA-6 cathode catalyst shown in Fig. 2.1.1.1-2.

The compatibility of acid fuel cell stacks with operation on medium-Btu gaseous fuels is presented in Fig. 2.1.1.1-3. This data provides design analysis information useful for the evaluation of typical fuels available from coal gasification and the impact that such fuel compositions will have on power plant operation.

Another important cell characteristic is fuel and air utilization. Typical curves of reactant and air utilization have been shown in the previous section for both one- and two-pass air flow. Fuel utilization in the mid to high 80 percent range appears feasible. The FCG-1 demonstration unit has a target of 83 percent fuel utilization.

Of particular importance to the integration of the cell with a gasified coal fuel supply system is the tolerance of the cell to impurities in the hydrogen feed. Carbon monoxide, which is a major product of most gasification processes, acts as a poison to the platinum catalyst. The mechanism is believed to be that of a simple site-elimination and results in decreased activity at the anode. The effect on performance is shown in Fig. 2.1.1.1-4 (Ref. 2.1-2). Also shown is the beneficial effect of increased temperature in improving the cell tolerance to CO. It is clear that a relatively efficient shift conversion from CO to H_2 will be required not only to improve fuel utilization, but to avoid cell degradation.

Other carbon compounds, CO_2 and CH_4 , common in gasifier off-gas streams, are reported not to be harmful and act only as a diluent. However, as with CO, the methane does represent a performance loss due to incomplete fuel utilization and if present in significant amounts would warrant special consideration.

The effect of sulfur compounds (H_2S and COS) on the cell is not well defined. The current approach for fuels containing more than .05 ppm sulfur is to catalytically hydrogenate the fuel, forming H_2S followed by absorption of the H_2S in

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FIG. 2.1.1.1-1

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INITIAL PERFORMANCE OF 20-CELL, 3.7-FT, RIBBED-SUBSTRATE STACK

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FIG. 2.1.1.1-2



PERFORMANCE HISTORY OF 20-CELL, 3.7-FT², RIBBED-SUBSTRATE DEVELOPMENT STACK

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FUEL EFFECT ON PERFORMANCE FOR 20-CELL, 3.7-FT? RIBBED-SUBSTRATE STACK

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FIG. 2.1.1.1-4



EFFECT OF CO CONTENT AND TEMPERATURE ON ANODE PERFORMANCE 0.35 mg Pt/CM²,269 mA/CM²

2.1-7

a bed of zinc oxide. As a result, there is virtually no sulfur in the feed and the catalysts in the fuel processing units and in the cell are not exposed to any significant amount of sulfur.

Subscale two-inch by two-inch cell tests have been carried out at UTC (Ref. 2.1-3) to determine the long term tolerance of phosphoric acid cells to the presence of H_2S in the fuel. One test cell was run for more than 2400 hours with 130 to 200 ppm H_2S added to the fuel. It showed long periods (900 hours) of stable operation followed by rapid performance decay. Performance could be restored by removal of CO from the fuel, removal of H_2S from the fuel, or by raising the cell temperature to 400 F.

The results of that testing are:

- 1. The long term tolerance of acid cells to H_2S in the fuel is not totally defined although over 2400 hours of operation on H_2S containing fuel has been demonstrated in a 2-inch x 2-inch cell.
- 2. The cell operated for periods up to 900 hours with 130-200 ppm $\rm H_2S$ in simulated reformer effluent with no decay.
- 3. Stable operation was followed by abrupt decay.
- 4. Post-test analyses of catalyst from a cell tested with H_2S added to the fuel showed no identifiable changes resulting from exposure to H_2S .

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Ammonia is a common constituent of the gasifier off-gas. It reacts with the phosphoric acid electrolyte to form (NH_4) H_2PO_4 , ammonium dihydrogen phosphate (ADHP). The formation of the ADHP results in a performance loss, possibly by changing the conductivity of the acid or by reducing the activity of the platinum by covering the active sites. Electrolyte conversion must be limited to less than about 0.2 mole percent to avoid unacceptable performance penalties. Diagnostic data on the cells tested indicate that most of the performance loss occurs at the cathode. Approximately 84 percent of the cathode activity is lost by conversion of only 1 percent of the acid to ADHP.

While the consequences of ammonia in the fuel gas appear severe, the resultant performance degradation is not irreversible. Long term tests have shown that the ADHP levels reach a constant value much lower than would be predicted by inlet ammonia concentrations. This, plus the fact that performance is restored when ammonia is removed from the feed, led to the hypothesis that the ammonium ion is oxidized at cell operating conditions. This has been confirmed by measurement of the current associated with the oxidation (Ref. 2.1-4).

In the design of a power system, each of the above characteristics must be considered and the optimum chosen on the basis of not only performance, but other important factors such as reliability and cost.

2.1-8

2.1.1.2 Molten Carbonate Cell Characteristics

A number of development programs are currently in progress or have already been reported on in the development of molten carbonate cell technology (Refs. 2.1-3 through 2.1-11). The questions concerning cell performance and cell life that remain to be investigated are summarized in Ref. 2.1-12.

These problem areas include loss of electrolyte, failure of the wet seal with reactant leakage, effect of structural changes on performance, cost-effective manufacturing techniques and telerance to impurities such as sulfur and chlorine. The last of these is an important factor in the integration of the fuel cell with a coal gasification and fuel gas cleanup system. While the tolerance to these impurities has not been fully defined, present studies are based on the premise that sulfur removal must be down to a level of less than 1 ppm. In general, this is most economically achieved by a combination of a regenerable removal system followed by a zinc oxide bed.

The basic cell design has been discussed previously. The overall cell reaction is simply the combination of one mole of hydrogen with one-half mole oxygen to produce water, electrical power and heat. Unlike the acid cell, carbon monoxide is not detrimental to performance and, in fact, is utilized by the cell since the water gas shift reaction is catalyzed by the anode $(CO + H_2O \ddagger CO_2 + H_2)$. The catalytic qualities are not all beneficial however, and under certain conditions solid carbon may be formed $(2CO \ddagger CO_2 + C)$. Additionally, methane which acts as an inert gas in the cell may be formed $(CO + 3H_2 \ddagger CH_4 + H_2O)$. Methane formation is harmful to performance since it passes through the cell without reacting.

Both carbon deposition and methane formation are enhanced by increased pressure. Thus, while high pressure operation is desirable from a performance standpoint, Fig. 2.1.1.2-1 (Ref. 2.1-13), care must be taken to eliminate the possibility of carbon formation and minimize methane formation. The humidification of the fuel gas stream has been found to be an effective means of helping achieve both goals.

The addition of water reduces the concentration of CO due to the water gas shift reaction. This is turn, reduces the likelihood of both carbon formation. Methane formation is controlled by the reduced CO concentration and the increased water vapor concentration. Cell operating pressure appears to be optimum at approximately 10 atm as will be discussed in the following section.

Cell operating temperature is based on a trade-off between ideal cell voltage, cell polarization and endurance. The electrolyte, which is comprised of a mixture of alkali metal carbonates in a ceramic matrix, is a solid at room temperature. The cell must be heated above the electrolyte melt temperature to provide the necessary ionic mobility to sustain cell reactions. As the temperature is increased beyond the melt point, the ideal cell voltage drops but the ionic mobility increases resulting in reduced polarization. The operating temperature range of the cell will be between 1100 and 1300°F (866 and 978°K); this permits waste heat to be removed as sensible heat in the cathode gas stream. A curve showing the composite effect of the variables is shown in Fig. 2.1.1.2-2 (Ref. 2.1-13). The knee of the curve, or point of diminishing return occurs at approximately 1200 F, which is the nominal cell operating temperature.

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EFFECT OF PRESSURE ON MOLTEN CARBONATE FUEL CELL PERFORMANCE 650°C, 160 mA/CM², 70 MOL % AIR/30 % MOL CO₂ OXIDANT AT 50% UTILIZATION

75% FUEL UTILIZATION

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FIG. 2.1.1.2-2



EFFECT OF TEMPERATURE ON MOLTEN CARBONATE FUEL CELL PERFORMANCE 1 atm, 200 mA/cm², 70 mol % AIR/30 mol % CO₂ OXIDANT AT 50% UTILIZATION 75% FUEL UTILIZATION

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In addition to operating temperature and pressure, three other major parameters define cell operation characteristics. These parameters include the fuel and oxidant utilizations and the cell performance. Utilization is defined as the ratio of the reactant consumed by the cell to the reactant supplied to the cell. Reactant utilizations determine the variation in reactant partial pressure over the cell and therefore the ideal cell voltage and driving forces for reactant diffusion at each point in the cell. This is an important determinant of cell performance. Because CO is shifted to H₂ and consumed in the molten carbonate cell, fuel utilization is defined as the ratio of the H₂ consumed in the cell to the total H₂ plus CO supplied in the cell.

Fuel and oxidant utilization are subject to optimization on a system level. Cell performance is a measure of the degree of development of the carbonate cell. The unit of performance for fuel cells is the design power density per square foot of active cell area. Cell power density is the product of the voltage measured across the electrodes of each cell and the current density of the cell at that voltage. This unit of performance is analogous to the horsepower to weight ratio for gas turbines. Since the capital cost of the cell stack is proportional to total cell area, high power densities provide lower cost.

At the time of the ECAS study (Ref. 2.1-14), cell performance was estimated to be as shown in Fig. 2.1.1.2-3. For the current DOE development program the cell performance goal is shown in Fig. 2.1.1.2-4. A comparison shows that the slope of the curve has been reduced to approximately midway between the 1976 data and that projected for a demonstration unit in the ECAS study. Table 2.1.1.2-1 gives the operating conditions for the current program.

As mentioned earlier, fuel and oxidant utilization is a subject for system integration. For the carbonate cell, the effect of reactant partial pressures on cell performance is difficult to portray because of the participation of CO at the anode due to the water gas shift reaction and consumption of both O_2 and CO_2 at the cathode. The effect of fuel utilization is shown in Fig. 2.1.1.2-5 (Ref. 2.1-5). The composite effect of fuel and oxidant utilization is shown in Fig. 2.1.1.2-6 (Ref. 2.1-13). Because of the number of variables involved, both fuel and oxidant side compositions, these curves are only tpical and a cell model is necessary to conduct the necessary system optimization.

The effects of impurities on the carbonate cell are not completely defined. Because sulfur is harmful in very low (\approx 1 ppm) concentrations, designs based on a current understanding of carbonate cell technology must reduce sulfur concentrations to that level. In the process, acids such as HCl that are known to be harmful to the cathode will be removed by a water wash prior to low-temperature desulfurization.

As with the acid cell, careful integration of the cell and other system components is necessary to produce an attractive system. Aug. 1997



ECAS CELL PERFORMANCE

FIG. 2.1.1.2-4

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MOLTEN CARBONATE CELL DESIGN PERFORMANCE

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TABLE 2.1.1.2-1

Stream	Anode		Cathode	
	Inlet	Exit	Inlet	Exit
Total Flow (Wet)				
Poun is	0.79	1.97	14.8	13.6
Mole	0.043	0.062	0.51	0.48
Composition (vol% dry)				
H ₂	38.3	5.57		
cō	56.0	3.84		
СН4	0.1	0.07		
co ₂	4.83	90.1	19.6	15.7
N ₂	0.7	0.4	70.5	76.3
02			9.9	8.0
H ₂ 0/Dry Gas (vol/vol)	0.755	0.702	0.27	0.29
Temperature (^O F)	1100	1220	1100	1300
Pressure (psia)	150	149	150	149
Utilization	85%			25%

CELL STACK FLOW CONDITIONS AT RATED POWER

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CELL PERFORMANCE AS A FUNCTION OF FUEL UTILIZATION

FIG. 2.1.1.2-6



EFFECT OF REACTANT UTILIZATION ON MOLTEN CARBONATE FUEL CELL PERFORMANCE 1 Atm, 200 mA/CM², 70 Mol % AIR/30 Mol % CO₂ OXIDANT SIMULATED REFORMED METHANE FUEL

2.1.2 System/Subsystem Designs

In this section the potential variations in system design and the optimization of the operating parameters are discussed. As mentioned earlier, because of its low operating temperature and requirement for operation on hydrogen alone, the acid cell offers little if any opportunity for variation in its configuration regardless of the gas supply. Therefore discussion in this section is limited to the carbonate cell. Work done at UTC (Ref. 2.1-9) formed the basis for that discussion. For the acid cell, the FCG-1 is typical of future designs and is described in Section 2.1.3.

Integration studies focused on the assessment of the impact on power plant efficiency of the following fuel cell configuration and operating parameters:

- . Cell Operating Pressure
- . Carbon Formation Criteria
- . Cell Cooling Mode
- . Cell Performance Technology Level
- . Fuel Utilization
- . Cell Stack Process Air Coolant Temperature Gradient
- . Cooling Recycle Loop Pressure Drop

These studies indicate trends in power section subsystem design options for further study with different gasifiers, and provide the basis for selection of a baseline power section system concept and a baseline set of cell operational parameters.

The power block is defined to include the fuel cell, bottoming cycle, and other conversion and associated equipment downstream of fuel gas cleanup. For these studies, the power block was assumed to be coupled with an oxygen-blown entrainedflow gasifier with physical solvent acid gas cleanup and zinc oxide solfur polishing similar to that described in Ref. 2.1-15. The fuel gas quality available to the fuel cell, the gasification section operating conditions, and steam and electrical requirements are also based on that study. The schematic of the power plant configuration is shown in Fig. 2.1.2-1. While the basic configuration uses a cathode recycle for heat removal, the anode recycle configuration option is shown as a dashed line. The steam bottoming cycle is not depicted; however, the schematic indicates locations of heat exchangers providing steam to the bottoming cycle or for process reheat. Care was taken to assure that the heat was of sufficient quality for high pressure steam generation, superheating, and/or steam reheat.

Important parameters impacting power plant performance and selection of the baseline power plant system concept are the fuel cell operating pressure, carbon formation criteria, and fuel cell cooling mode. Both anode recycle and fuel gas steam resaturation are design options that were studied to prevent carbon deposition. Three cooling modes were considered: cathode recycle cooling (RR = 0), anode recycle cooling (no cathode recycle), and dual recycle cooling utilizing small anode recycles to reduce the fuel gas resaturation requirements.

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GAS FLOW SCHEMATIC

Initial studies focused on design requirements to prevent carbon deposition over a range of pressures. Based on UTC cell tests and literature data, full C-H-O fuel gas equilibrium has been assumed for the carbon laydown criterion; i.e., at a given temperature and pressure carbon (graphite) deposition will not occur unless thermodynamically favored based on shift and reform equilibria at those conditions. However, for purposes of assessing the impact associated with more stringent design requirements, a second carbon deposition criterion was also investigated. This criterion, defined as partial C-H-O fuel gas equilibrium, assumes carbon (graphite) deposition will not occur unless favored thermodynamically based on shift equilibrium only (frozen methane).

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Figure 2.1.2-2 indicates the impact on overall power plant efficiency of the three parameters. The results are normalized for constant cell power density (implies constant \$/kW gross dc power) utilizing a performance model with a technology projection similar to that in the ECAS (Ref. 2.1-14) study (Fig. 2.1.1.2-3).

Trends indicated by these parameter studies are as follows:

- . Maximum system efficiency is at about 150 psia irrespective of carbon criterion and cooling mode, however, higher pressures have only a small negative impact on efficiency, based on the full C-H-O equilibrium carbon criterion.
- . Carbon criterion has a significant impact on efficiency, especially at high pressures.
- . Carbon criterion defines the preferred cooling mode. With full C-H-O equilibrium, water resaturation with cathode recycle cooling is the preferred approach, while with partial C-H-O equilibrium, anode cooling is indicated.

Another important parameter influencing the power plant efficiency is the cell performance technology level. Assuming a nearer-term level of performance results in a significant drop in efficiency, as illustrated in Fig. 2.1.2-3. Power plant efficiency again maximizes at \sim 150 psia, independent of cell reformance level.

The above studies assumed the following operational power section parameters to be constant:

- . Fuel utilization = 85%
- . Cell stack process coolant temperature gradient (ΔT) = 200°F
- . Recycle pressure drop $(\Delta P) = 5$ psia

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FUEL CELL OPERATING PRESSURE EFFECT ON ESTIMATED POWER PLANT EFFICIENCY

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FUEL CELL PRESSURE AND PERFORMANCE LEVEL EFFECT ON OVERALL POWER PLANT EFFICIENCY

The impact on efficiency of varying the above parameters are shown in Figs. 2.1.2-4, 2.1.2-5 and 2.1.2-6 respectively. The following summarizes the results of these operational parameter studies.

- . Fuel gas utilization has a significant impact on system efficiency. System efficiency increases at high fuel utilization due to the higher efficiency of the fuel cell as compared to the (steam) bottoming cycle.
- . Higher stack temperature gradients for cooling increase system efficiency due to reduced compressor parasite requirements.
- . Lower recycle pressure drops increase efficiency for the same reason. Alternatively, higher pressure drops increase optimum cell pressure; however, higher ΔP 's tend to flatten the optimum peak, making power plant efficiency insensitive to operating pressure in the high range.

2.1.3 Description of System Baseline Technology

In this section, the systems representative of both the phosphoric acid cell technology and molten carbonate cell technology are described. The acid cell description is based on existing technology while further development is required for the carbonate cell as described in previous sections.

2.1.3.1 Phosphoric Acid Cell Technology

The 4.8-MW fuel cell demonstration unit, while not a commercial prototype, is representative of future plants. To enter widespread commercial service, it will be necessary to reduce the cost of the powerplant and increase its durability. At its inception, the 4.8-MW demonstrator was one module of the larger FCG-1 power plant. Currently, while similar in layout and operation, the FCG-1 design is based on more advanced cell technology and is designed for higher temperature and pressure operation.

In improving operating conditions, current density has been kept constant and efficiency has been improved. The difference in operating conditions was shown in Fig. 2.1.1.1-1. Current cells operate at low (3 atm) pressure and nominally 375 F. Advanced cells, such as would be used in a commercial plant would operate at approximately 6 atm and 400 F. A specification for the FCG-1 is given in Ref. 2.1-16.

The dc module is the heart of the power plant. Its purpose is to convert hydrocarbon fuel and air into direct current. The major functional components of the dc module are shown schematically in Fig. 2.1.2-7. The module consists of fuel and air processing subsystems shown at the left in the figure, a fuel cell power section in the center, and a thermal management subsystem at the upper right. For simplicity, the schematic omits a number of valves and heat exchangers needed for flow and temperature control.

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OVERALL H2 UTILIZATION, PERCENT

OVERALL H2 UTILIZATION EFFECT ON OVERALL POWER PLANT EFFICIENCY

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CELL STACK TEMPERATURE GRADIENT EFFECT ON OVERALL POWER PLANT EFFICIENCY



RECYCLE PRESSURE DROP AND CELL OPERATING PRESSURE EFFECT ON OVERALL POWER PLANT EFFICIENCY

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DC MODULE SYSTEM SCHEMATIC



During demonstrator module operation, air enters one of the compressors, shown at the upper left in the schematic, and is compressed to 35 psig. The lowflow turbocompressor will normally be used below 35 percent of rated power, and the high-flow unit at all other powers. This pressurized air stream is then divided by valves and controls into three streams. The main stream is directed to the cell cathodes to supply the oxygen required for the fuel cell reaction. The fuel cell reaction produces heat and water. Most of this product water evaporates into the air stream flowing through the cathode gas space. This hot, moist, cathode-effluent stream then passes through a regenerator and a liquidcooled condenser where it is cooled sufficiently to condense the water required for use in the fuel processing subsystem. The cathode exhaust stream, combined with the hot reformer-burner effluent, provides the energy to drive the turbine of the turbocompressor unit.

Power plant process fuel is sometimes burned in the auxiliary burner to add energy to the gas stream before it enters the turbine. The system components are matched such that, at power plant rated output power, the energy in the gas stream is sufficient to power the turbine. As the powerplant output level is reduced, the energy available in the gas stream decreases. Since the turbine requires a near-constant energy input as the output power is reduced, process fuel is burned to make up the small energy deficit at part power.

At the bottom left of the schematic, naphtha fuel, which has been pressurized by a pump, is mixed with hydrogen-rich gas from the fuel processor exit stream. The mixture passes through a hydrotreater (hydrodesulfurizer) where sulfur compounds are converted to hydrogen sulfide and absorbed in a zinc oxide bed. Any halogens in the fuel are absorbed in a guard bed. The clean fuel is then mixed with steam made by power section waste heat. The fuel-steam mixture passes into the reformer. There it is processed to a mixture of hydrogen, carbon dioxide, carbon monoxide, and water vapor. The carbon monoxide and water are then further converted to hydrogen and carbon dioxide in two shift converters operating at successively lower temperatures. The low temperature shift converter reduces the carbon monoxide concentration to less than one percent. The processed fuel next flows to the anode cavities in the fuel cell power section where hydrogen is extracted for the electrochemical reaction. The anode effluent gas stream passes through a regenerator and condenser where it is cooled to condense additional water for power section cooling and steam generation. The anode exhaust stream then enters the reformer burner where the dilute hydrogen is burned to provide heat for the reforming process. Then, as described above, the burner gas stream is combined with the cathode exhaust stream before entering the turbine.

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In the power section, cells are stacked in a series-parallel arrangement to produce 3000-volt direct current which is fed to the power conditioner. Power section temperature is controlled by a thermal management subsystem. Steam is generated from condensed water using power section waste heat. As shown at the upper right quadrant of the schematic, cooling water enters the fuel cell coolers, is partially converted to steam, and continues on to the separator where the steam is removed. Part of

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the steam is sent to the reformer; the remainder is condensed to maintain the power section at the proper temperature. Water from storage is introduced into the thermal management loop to maintain its inventory. The cooling water is pumped back to the power section. A heat exchanger combined with a burner serves as a source of heat during powerplant startup and standby operation. Excess heat from the dc module is rejected to the dry air cooling towers via a liquid loop. This heat could also be used to provide steam for industrial processes or hot water for space heating or washing.

2.1.3.2 Molten-Carbonate Fuel Cell Technology

A schematic of a molten-carbonate power system developed by UTC for use with an oxygen-blown Texaco gasifier is shown in Fig. 2.1.3.2-1. A combination of Selexol cleanup followed by a zinc oxide bed is used to reduce sulfur to tolerable levels. The system was fully integrated and gave an overall efficiency of 47.9% from coal (HHV) to electricity.

Clean fuel gas enters a Fuel Gas Expander 50-1-EX-2, which drives the Anode Recycle Compressor 50-1-C-2. The fuel gas serves as makeup gas to the anode recycle loop at the discharge of the Anode Recycle Compressor 50-1-C-2. The anode recycle gas stream continuously circulates through the anode side of the 50 parallel fuel cells to supply fuel and to remove the heat of reaction. The 1300°F anode recycle exhaust gas from the fuel cells is cooled in the two pressurized Heat Recovery and Steam Generation units connected in series. The first unit contains Steam Reheater 51-1-E-4 and Superheater 51-1-E-3 as internal coils. The second unit contains HP Boiler 51-1-E-2 and BFW Preheater 51-1-E-1 as internal coils. The cooled anode recycle gas then enters the Anode Recycle Compressor 501-1-C-2 at 1096°F for recompression and recycle. This whole anode recycle loop is designed to operate with a total pressure drop of 5.4 psi.

A portion of the anode effluent at 1300°F combines with inlet air and enters Catalytic Burners 51-1-V-1A&B for combustion. The exit flue gas then flows to the cathode side of the 50 parallel fuel cells. The cathode exhaust gas flows to Exhaust Gas Expander 50-1-EX-1, which drives Air Compressors 50-1-C-1 and 11-1-C-1, and Generator 50-1-G-1.

The gas stream from Exhaust Gas Expander 50-1-EX-1 flows through several economizers including BFW Preheater 51-1-E-5 and Condensate Preheater 51-1-E-7 for recovery of waste heat before discharging to the atmosphere at 250° F.

System performance is summarized in Table 2.1.3.2-1. The nominal cell conditions selected for this study were 150 psia and 1200°F. In this study, process fuel gas cooling was selected to provide for fuel cell thermal management and transfer the rejected heat of reaction to the bottoming cycle via an anode recycle loop. This approach assures that the fuel gases entering the anode are thermodynamically stable with regard to methanation and carbon deposition. However, the kinetics of these reactions may permit less stringent design criteria than was assumed in this study.



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TABLE 2.1.3.2-1

SUMMARY OF SYSTEM PERFORMANCE FOR OXYGEN-BLOWN TEXACO GASIFIER/MOLTEN-CARBONATE FUEL CELL

Gasification And Gas Cleaning System		
Coal Feed Rate, lbs/hr (m.f.)	798,333	
Oxygen*/Coal Ratio, lbs/lb m.f.	0.	858
Oxidant Temperature, °F	300	
Steam/Coal Ratio, lbs/lb m.f.	0	
Slurry Water/Coal Ratio, lbs/lb m.f.	· 0.	503
Gasification Section Average Pressure, psig	600	•
Crude Gas Temperature, °P	2,300-2,6	00
Crude Gas HHV (dry basis), Btu/SCF**	281.	1
Temperature of Fuel Gas to Fuel Cell Loop °F	738	
Temperature of Fuel Gas to Zinc Oxide Beds	700	
Gaseous Sulfur Emission, 1b SO ₂ Equivalent/10 ⁶ Btu Coal	1	0.009
Power System		
Air Compression Pressure Ratio (50-1-C-1)	10.4:1	
Fuel Cell Inlet Temperature, °F	1,100	
Fuel Cell Exhaust Temperature, °F	1,300	
Steam Conditions, psia/°F/°F	1800/1000/1	000
Condenser Pressure, Inches Hg abs	2.	5
Anode Recycle Loop Pressure Drop, psi	5.	4
Stack Temperature, *F	250	
Fuel Cell Power#, MW	1,002.	8
Steam Turbine Power#, MW	49 7.	6
Exhaust Expansion Turbine Power#, MW	40.	4
Total Power Consumed, MW	110.	8
Net System Power, NW	1,430.	0
Overall System		
Process and Deaerator Makeup Water, gpm/1000 MW	445	
Cooling Tower Makeup Water, gpm/1000 MW	4,480	
Cooling Water Circulation Rate, gpm/HW	208	
Cooling Tower Heat Rejection, % of Coal HHV	29.	0
Air Cooler Heat Rejection, 🕏 of Coal HHV	7.	6
Net Heat Rate, Btu/kWh	7,130	
Overall System Efficiency (Coal + Power), % of Coal HMR	7 4 7.	9

*Dry basis, 100 percent 02

**Excluding the HHV of H_2S , COS and NH_3

#At generator terminals

As discussed in the previous section, improved performance is possible with steam injection and cathode recycle. In addition to operating temperature and pressure, other major parameters defining cell performance and efficiency are cell voltage, and the fuel and oxidant utilizations. Utilizations define the ratio of the reactant consumed by the cell to the reactant supplied to the cell. For a given set of reactant compositions, utilization determines the reactant partial pressure and therefore the maximum ideal cell voltage. Because CO is shifted to H, and consumed in the molten carbonate cell, fuel utilization is defined as the ratio of the H₂ consumed in the cell to the total H₂ plus CO supplied to the cell. Fuel utilization and cell voltage are measures of prime cycle system efficiency since they indicate the percentage of CO and H₂ in the fuel gas that is consumed electrochemically and the efficiency of that electrochemical conversion process. A preliminary cost and efficiency trade-off study set the cell voltage at 0.75 volts per cell, and overall fuel utilization at 90 percent. The oxidant air flow rate was set to maintain a two to one ratio between carbon dioxide and oxygen partial pressures on the cathode, a value that optimizes cathode performance. A summary of fuel cell design parameters is given in Table 2.1.3.2-2.

Other cell design conditions that affect overall power plant characteristics are heat losses and cell operating life. The cells are thermally insulated and cell stack heat losses were calculated to be two percent of the design gross power output. For purposes of defining base case economics, cell useful life was assumed to be 40,000 operating hours. This is a goal that has been set for the nearer term acid cells and appears to be a reasonable goal for molten carbonate cells as well. It should be pointed out that after 40,000 hours of operation, the cell with continue to operate, however, at a slightly reduced performance level than assumed for this study. Thus, a utility may choose to replace the fuel cells at this time or it may choose to continue to run the plant with the existing cells, but at a somewhat reduced power level or efficiency.

The fuel cell module produces dc power which is converted to ac in the inverter system. The inverter operates at 96 percent efficiency at rated load, and outputs 60 Hz, 3-phase ac power. Included in Table 2.1.3.2-2 is the power output summary from the inverters. The inverters selected utilize solid-state, self-commutated technology. An important characteristic of utilizing self-commutated technology is that both real and reactive power can be dispatched. Thus, the power factor may be operated over the range of 0.0 to 1.0 within the MVA rating of the inverters, precluding the need for power factor correction capacitors.

The inverters are modular and are designed to be factory fabricated and rail transportable. Each inverter module consists of three inverter bridges connected in parallel to the dc bus. Output voltage harmonics up to 17 times the fundamental are cancelled in harmonic reduction transformers. Any single harmonic voltage кð1-955343-9

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TABLE 2.1.3.2-2

FUEL CELL DESIGN PARAMETERS

Cell Voltage ~ Volts	0.75
Fuel Utilization ~ 💲 *	50
Oxidant Utilization ~ %	58
Gross dc Power Density ~ Watts/Ft ²	148

DC Module Operating Characteristics

Dc Hodule Output ~ MW	4.18
Number of dc Modules	250
Total dc Output to Inverter ~ MW	1044.6

*Per pass utilization includes anode recycles. Overall utilization for each case is 90 percent

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output is limited to less than one percent of the fundamental. Series reactors are inserted between the bridge output and the harmonic cancelling transformers of each inverter module to provide an inductive impedance to the utility line for control purposes and for buffering the bridge from line transients. The 60 Hz, three-phase output from the inverter modules is then combined in an inverter transformer for step-up to an intermediate ac paralleling voltage.

The remaining components in the prime cycle system are the catalytic burners and the turbocompressors and auxiliary generator equipment. The catalytic burners oxidize hydrogen, carbon monoxide, and methane in the fuel anode vent. Adiabatic oxidation temperature is quite low, approximately 1158 F. Kinetic studies indicate that catalytic combustion in these temperature regimes yields very low nitrogen oxide and carbon monoxide emission levels. The catalytic burners utilize a precious metal catalyst supported on a ceramic material, and are enclosed in insulated carbon steel pressure vessles. Design life for the units is two years with replacement of the catalyst bed after this period.

2.1-34

2.1.4 References

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- 2.1-11 General Electric Co.: Development of Molten Carbonate Fuel Cell Power Plant. Quarterly Progress Report, DOE/ET/17019-1, April 1980.
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- 2.1-13 Institute of Gas Technology: Handbook of Fuel Cell Performance. DOE-COO-1545-T1, May 1980.
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- 2.1.4 References (Cont'd)
- 2.1-15 Fluor Engineers and Constructors, Inc.: An Economic Comparison of Molten Carbonate Fuel Cells and Gas Turbines in Coal Gasification-Based Power Plants. EPRI Report No. AP-1543, September 1980.

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3.1 Evaluation of Coal-Derived Fuels for Advanced Fuel Cell Power Systems

The objective of this task is to determine the overall power plant efficiency of a fuel cell system using medium-Btu gas delivered to the power plant via pipeline. The properties of the gas delivered to the power plant will be obtained and any further processing, if necessary, will be identified. Based upon the fuel gas characteristics, the performance of the fuel cell power plant will be described.

3.1.1 Definition of Fuel Gas Requirements

As discussed in Sections 1 & 2 both the phosphoric acid and molten carbonate type cells are susceptible to degradation from impurities in the fuel. Preliminary fuel specifications have been established for the FCG-1 Power Plant (Ref. 3.1-1). These are presented in Tables 3.1.1-1, 2 & 3. The gaseous fuels of interest here are covered by Table 3.1.1-1. While that specification allows a sulfur level of 100 ppm, the system includes a zinc oxide bed to achieve nearly complete sulfur removal prior to entering the reformer and water gas shift sections of these fuel processing system. While the acid cell can tolerate some H_2S , it appears to have a synergistic effect when accompanied by CO. Since it will not be possible to shift all the CO to H_2 , it appears prudent to remove all the sulfur even if catalyst poisoning could be overcome.

In the case of the carbonate cell, allowable sulfur levels have yet to be established. However, it appears likely that these levels will be less than 1 ppm. Cell tolerance to other impurities have been estimated (Ref. 3.1-2) and are presented in Table 3.1.1-4. For purpose of this study it was assumed that low levels (<1%) of C_2 hydrocarbons would act as a diluent and not affect cell performance.

The oxygen-blown Texaco and British Gas Corp. (BGC) slagging gasifiers were selected as the gasifiers to provide medium-Btu fuel gas for transport in pipeline to the power generation site. Properties of the fuel gas from each of these gasifiers, based upon Ref. 3.1-3, are given in Table 3.1.1-5. The total sulfur content of '00 ppm H_2S + COS is that agreed upon by NASA and the Contractor (Ref. 3.1-4) as being typical of pipeline quality fuel gas.

TABLE 3.1.1-1

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GASEOUS FUEL SPECIFICATION

Gaseous fuels shall consist of various combinations of saturated hydrocarbon compounds, hydrogen, carbon monoxide and incrts, with the following chemical composition limits:

COMPOSITION LIMITS

Impurity	Upper Limit
Sulfur, ppm vol	100
Nitrogen, vol %	10.0*
Oxygen, vol %	0.1
Olefins, vol %	3.0

* Limit applies only for gases containing hydrocarbons.

Acceptable gaseous fuels include:

Coal Gases from Entrained Flow, Fluidized Bed, or Moving Bed Processes. Natural Gas Synthetic Natural Gas Propane Hydrogen Anaerobic Digestor Gases Some Industrial Byproducts

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LIQUID FUEL SPECIFICATION

Liquid fuels shall consist of saturated hydrocarbon compounds with the following chemical composition and physical property limits:

COMPOSITION LIMITS

Impurity	Upper Limit
Aromatics, vol %	12.0
Olefins, vol %	3.0
Sulfur, ppm wt.	500.0
Chlorine, ppm wt.	3.0
Lead, ppm wt.	1.0

PHYSICAL PROPERTY LIMITS

Reid Vapor Pressure, Maximum, 1b 12.5 Final Boiling Point, Maximum, ^oF 420

Acceptable liquid fuels include:

Light Petroleum Distillates Light Coal Liquids

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TABLE 3.1.1-3

ALCOHOL FUEL SPECIFICATION

Alcohol fuels shall consist primarily of methyl, ethyl, isopropyl and isobutyl alcohol, or mixtures thereof, with the following chemical composition limits:

COMPOSITION LIMITS

Impurity	Upper Limit
Sulfur, ppm wt.	0.1
Chlorine, ppm wt.	0.1
Lead, ppm wt.	1.0

Methyl and isopropyl alcohols may be used as denaturants.

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TABLE 3.1.1-4

ESTIMATED MAXIMUM ALLOWABLE IMPURITY LEVELS FOR PHOSPHORIC ACID FUEL CELLS

Impurity	<u>Maximum Limit</u>
co ₂	diluent
CH4	diluent
N ₂	diluent
co	1%
H ₂ S, COS	100 ppm
c ₂ +	100 ppm
C1-	1 ppm
NH3	1 ppm
Metal ions (Fe, Cu, etc.)	nil

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TABLE 3.1.1-5

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GAS COMPOSITIONS

	Texaco Gas	BGC Gas
Element	Vol. %	Vol. X
СН	0.1	8.4
H ₂	35.6	30.6
cō	52.4	58.5
co ₂	10.8	1.9
$H_2\bar{S}$)		
∓ }	100 ppm	100 ppm
cos)		•.•
N ₂	0.8	0.5
Ar	0.2	0
H ₂ 0	0.1	0.1
-	100.0	100.0

Heating Values, LHV

LHV, Btu/scf	267.5	349.1
LHV, Btu/1b	4949	6825

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Fuel delivery pressure is dictated by the combined cycle of Section 3.2 in which the combustor operates at 175 psia. Allowing for fuel handling, a pressure of at least 250 psia would be required at delivery. It is anticipated that the Texaco and BGC gasifiers would be operated at pressures well beyond this value. Since the fuel cells operate at lower pressures it may be possible to recover some of the compressor work with an expander.

Fuel delivery temperature is not critical unless it is low enough to allow condensation. For the moisture content given in Table 3.1.1-5, temperatures should be above 75 F. An average temperature of 86 F has been assumed.

The required flow rate of fuel gas is a function of heating value and output power levels. For the carbonate systems, flows were selected that provided a heat input comparable to the integrated systems which were sized for a nominal 500 MW output. The flow rate is 527,310 lb/hr for the Texaco gas and 453,190 lb/hr for the BGC gas. In the case of the phosphoric acid cells, flows were made to be compatible with the FCG-1 module.

3.1.2 - Identification of Power Generation Facility

3.1.2.1 Phosphoric Acid Cell

The basic generating system arrangement has been described in Section 2.1.3. The plant, while initially intended for use on light distillate fuel, is designed for fuel flexibility. The fuel processor can be adapted to various gaseous and liquid fuels including those under consideration here.

For operation on naptha, a system schematic and flow sheet are shown in Fig. 3.1.2.1-1 and Table 3.1.2.1-1 (Ref. 3.1-5). For use with either of the gases being considered here, minor changes are required. These will be discussed using the naptha fueled plant as a baseline to describe fuel processor and fuel cell operation.

Referring to Fig. 3.1.2.1-1, naptha with no sulfur (Stream 1) enters the system and passes through the hydrodesulfurizer and zinc oxide bed prior to entering the reformer.

Fuel containing sulfur would be mixed with hydrogen rich gas prior to entering the hydrodesulfurizer. With the medium-Btu gases being considered here, the sulfur has already been combined with hydrogen and can be sent directly to the zinc oxide beds located just downstream of the hydrodesulfurizer. Removal of 100 ppm of sulfur from the gas puts a burden on the zinc oxide bed. The cost of desulfurization by the zinc oxide is on the order of 1 mill per kWh.

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TABLE 3.1.2.1-1

MASS AND ENERGY BALANCE FOR SYSTEM FOR PHOSPHORIC ACID FUEL POWER SYSTEM

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0.66	0.94	4.18	2.28	č. 30	2.24	8.22	1.21	5.48	0.85	3.47	1.09	huty, MMBtu/hr	Q
E-13	E-12	E-11	E-10	E-9	E-7	E-6	E-5	E-4	E-3	E-2	E-1	xchangers	ш
134	12,3	29,763	29,082	,858	Ŋ	10,608	80	10,6	8,851	1,356	250	. b/hr 2,:	-
8.	426	1970.5	1800.3	87.1	Ä	747.6	•6	747	491.7	ł) /	OTAL	Г
0	337	1422.3	1422.3			1		1	1	1		N2	1
8	89	207.9	378.0	ł		ł		1	ł	1	ł	02	
	I	ł	1	1.4		1.4	. 4	1	ł	ł	ł	CH4	
_	ł	340.3	1	26.4		246.7	.5	298	491.7	1	1	н ₂ 0	
	1	ł	ł	17.7	**	117.7	6.	65	ł	ł	1	c0 ₂	
	1	ł	ł	3.7		3.7	•6	55	ł	ł	ł	18	
	ł	ł	ł	37.8		378.0	.2	326	1	ł	ł	Н,	
												lomp ott en t	0
-	5	ø	7	9		Ś		4	ŝ	2	1	itream No.	S

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Following desulfurization, the gas is mixed with steam (Stream 3) and sent to the reformer where the hydrocarbons are converted to hydrogen along with some carbon monoxide, carbon dioxide and methane (Stream 4). These gases are cooled and sent to the shift converter where most of the CO along with a corresponding amount of H_2O is shifted to make H_2 and CO_2 (Stream 5). In the case of the Texaco gas, the reformer can be bypassed and the desulfurized gas sent directly to the shift converter. However, because of the high CO level (52%) in the gas it is necessary to add two high-temperature shift converters with associated heat exchangers to reduce the CO concentration to a level typical of a steam reformer product. The gas can then be processed in the normal shift converter. 1

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In the BGC gas, methane, while only 8% by volume, accounts for 23 percent of the heating value. Therefore, to avoid the loss of that energy, the BGC gas must be processed in the reformer. Because of the low methane concentration and the exothermic shift reactions that will also occur in the reformer, little heat is required. For the BGC gas, no change in the basic fuel processor arrangement is required. However, the size of the low-temperature shift converter must be increased.

After shifting, the fuel gas is cooled and most of the water vapor is condensed and removed for re-use in the reformer. It is then sent to the anode side of the cell. In the fuel cell, 10 percent of the hydrogen is consumed. The remainder, along with residual CO and CH_2 , is combined with incoming fuel to provide heat for the reformer.

A comparison of the performance of the phosphoric acid power plant using maptha, the Texaco gas and the BGC gas is given in Table 3.1.2.1-2. As can be seen, there is little change in either heat rate or the availability of low-grade heat as the system is changed from one fuel to the other.

The operation of the FCG-1 power plant on medium-Btu gas both with and without methane is the subject of a study being performed for TVA (Ref. 3.1-6). The final report of that work will not be available for several months. However, it will include a more detailed description of the operation of the cell than is possible at this time.

3.1.2.2 Molten Carbonate Cell

The basic generating system arrangement has been discussed in Section 2.1.3. The carbonate cell, by virtue of its high operating temperature is well suited for integrated operation with the gasifer. However, in the over-the-fence configuration it also exhibits a low heat rate. The system arrangement is shown in Fig. 3.1.2.2-1. It is the same for use with either the Texaco or BGC gas. Table 3.1.2-2 summarizes performance for both gases. For the Texaco gas, a heat and mass balance is given in Table 3.1.2.2-3 and stream compositions are given in Table 3.1.2.2-3. For the BGC gas, the corresponding tables are 3.1.2.2-4 and 3.1.2.2-5 respectively.

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TABLE 3.1.2.1-2

SUMMARY OF COAL-DERIVED FUEL IMPACT

11.3-MWAC Net Power Rating

Fuel	Texaco	BGC
Variation from Basic Power Plant	Add 2-Stage Shift Converter	Increase Low- Temperature Shift Converter Size
	Increase Low- Temperature Shift Converter Size	
Rated-Power Heat Rate, Btu/kWh	8350	8360
Recoverable Heat, 10 ⁶ Btu/h		
High Grade, STEAM	4.2	6.2
Low Grade, 160 F WATER	37	37
Overall Energy Utilization, (Based on HHV)	% 85	86

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TABLE 3.1.2.2-1

PERFORMANCE SUMMARY FOR MOLTEN CARBONATE POWER SYSTEMS AND MEDIUM-BTU FUEL GAS

Fuel Gas:		
Gasification Process	Texaco	BGC
Flow Rate, 1b/hr	527310	453190
Lower Heating Value, Btu/1b	4949	6826
Performance:		
Fuel Cell Power, MW	346.0	330.4
Gas Turbine Power, MW	100.3	107.5
Steam Turbine Power, MW	52.0	119.2
Utilities ⁽¹⁾ , MW	30.8	29.2
Total Net Power, MW	485.5	527.9
Heat Rate, Btu/kWh ⁽²⁾	5375	5860
Overall Efficiency, $\chi^{(2)}$	63.5	58.2

(1) Includes fuel compressors and 3% for plant requirements

(2) Based on lower heating value of gas.

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TABLE 3.1.2.2-2

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HEAT AND MASS BALANCE FOR MOLTEN CARBONATE POWER SYSTEM AND TEXACO FUEL CAS

Station Number	Temperature (°F)	Pressure (psia)	Flow (1b/sec) ⁽¹⁾
1	59	14.7	754.4
2	595	150.8	754.4
3	80	275.0	146.5
4	700	270.0	146.5
5	656	260.8	246.9
6	1100	256.1	246.9
7	1100	150.0	246.9
8	1220	150.0	568.4
9	981	147.5	568.4
10	809	145.0	568.4
11	820	150.8	568.4
12	988 •	150.0	1322.8
13	1100	150.0	3944.5
14	1300	150.0	3622.9
15	1300	150.0	1001.2
16	659	16.2	1001.2
17	480	14.7	1001.2
17A	363	14.7	1001.2
18	300	14.7	1001.2
19	1300	150.0	2621.7
20	1300	150.0	2162.0
21	1213	150.0	2162.0
22	1300	150.0	459.7
23	1213	150.0	459.7
24	1213	150.0	2621.7
25	1140	145.1	2621.7
26	1154	150.0	2621.7
27	108	1.23	159.8
27A	108	30.0	159.8
28	218	30.0	159.8
29	250	30.0	165.0
30	255	2837.0	165.0
31	446	2553.2	165.0
32	662	2400.0	165.0
33	662	2400.0	165.0
34	1000	2400.0	165.0

(1) All flow values are for one engine.

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TABLE 3.1.2.2-2 (Cont'd)

HEAT AND MASS BALANCE FOR MOLTEN CARBONATE POWER SYSTEM AND TEXACO FUEL GAS

35	653	600.0	64.6
36	653	600.0	100.4
37	611	260.8	100.4
38	1009	540.0	64.6
39	405	34.0	5.2
40	108	1.23	57.4

				MATERI POWER	AL BALANCE SYSTEM US	E FOR MOLT	en carbona o fuel cas	TE			
C	Number	-	2	£	-4		-1	œ	11-		12
	M.W.	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr
	16.04	ł	8	417	26	417	26	417	26	ł	ł
	2.016	İ	8	18507	9180	18507	9180	4167	2067	1	ł
	28.01	1	1	378499	13513	378499	13513	37477	1338	ł	1
	44.01	ł	ţ	122040	2773	122040	2773	1506814	34238	8	35601
	34.08	ł	ł	1	1	ł	1	ł	ł	ł	ł
	60.07	ł	ł	1		1	ł	1	ł	1	ł
	32.00	625696	19553	:		ł	1	1	1	569600	17800
	28.02	2036746	72689	5912	211	5912	211	5912	211	2042658	72900
	96. 6E	36465	619	1478	37	1478	37	1478	37	37943	950
	18.02	37247	2067	469	26	361896	20083	490090	27197	545375	30265
		2736154	95222	527322	25766	888749	45823	2046355	65114	4762376	157516

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TABLE 3.1.2.2-3

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TABLE 3.1.2.2-3 (Cont'd)

MATERIAL BALANCE FOR MOLTEN CARBONATE POWER SYSTEM USING TEXAGO FUEL GAS

										07 60	
				71		15-1	8	19-	-26		
Station	Number	13		4	•			-17,11-	Wale/Hr	Lb/Hr No!	ls/Hr
			-1-14-	Lb/Hr	Mols/Hr	Lh/Hr	bls/Hr	1N /0'J			
Comp.	М.Ч.	Lb/Hr F	111/910				2	ł	ł	:	1
		ļ	ł	1	1	1					
ਚੈਂ	16.04	1				1	ł	ł	1	1	ł
н Н	2.016	ł	1	ł	1			ļ	;	ł	1
7			1	ł	1	1	ł	•			
8	28.01	1			1003	717847	16311	1879667	42710	1	١
ຣິ	44.01	3446467	78311	2597514	17060				1	ł	1
7~~			1	ł	ł	ł	ł	•	I		
H2S	34.08	1	1			ļ	ł	1	1	ł	1
	1		ļ	1	1	•					
COS	60.07	1				76007	8156	683360	21355	•	1
Ċ	10 DO	1252960	39155	944352	29511	766007	 		00000	- 1	1
0 ₂	20.10		002536	7391396	5 263790	2042658	72900	5348738	190890	ł	1
N ₂	28.02	9461967	061007		0576	37943	950	11466	2489	1	ļ
~	39.94	137354	3439	13732	4		29000	142803]	79247	575433	31933
:	() a.	1973320	109510	197337	0 190510	545375	10705				51015
H20	70.01			0007000	16524	3604815	128582	943920	7 336691	51:433	<i>ccc</i> 10
Total		14201547	494205	1304390	1.3COt 00						

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TABLE 3.1.2.2-4

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Station Number	Temperature (°F)	<u>Pressure (psia</u>)	Flow (1b/sec) (1)
1	59	14.7	865.4
2	5 9 5	150.8	865.4
3	80	275.0	125.9
4	700	270.0	125.9
5	649	260.8	236.3
6	1100	256.1	236.3
7	1100	150.0	236.3
8	1220	150.0	532.1
9	969	147.5	532.1
10	808	145.0	532.1
11	819	150.8	532.1
12	1375	150.0	1397.6
13	1100	150.0	3759.9
14	1300	150.0	3464.1
15	1300	150.0	1101.8
16	655	16.2	1101.8
17	485	14.7	1101.8
18	300	14.7	1101.8
19	1300	150.0	2362.3
20	1300	150.0	1763.9
21	1115	150.0	1763 .9
22	1300	150.0	598.5
23	1115	150.0	598.5
24	1115	150.0	2362.3
25	9 24	145.1	2362.3
26	937	150.0	2362.3
27	108	1.23	251.0
28	108	30.0	251.0
29	250	30.0	286.3
30	255	2837.0	286.3
31	446	2553.0	286.3
32	602	2400.0	286.3
33	662	2400.0	286.3
34	1000	2400.0	286.3
35	648	600.0	175.9
36	648	600.0	110.4
37	604	260.8	110.4
38	1000	540.0	175.9
39	385	34.0	35.3
40	108	1.23	140.6

HEAT AND MASS BALANCE FOR MOLTEN CARBONATE POWER SYSTEM AND BGC FUEL GAS

(1) All flow values are for one engine.

R81-9	55343-9		/Hr				368			934	496	048	288	634
		~	Mols				33			16	83	-1	34	169
		11	Lb/Hr				1490531			541888	2339558	41857	617870	5031704
		11	Mols/Hr	1966	1983	1149	30753				105		37383	63239
		80	Lb/Hr	31535	3997	32183	1353440	ŧ	H H	ł	2942		491640	1915737
ĥ	CARBONATE UEL CAS	7	Mols/Hr	1966	7179	13699	456				105		22082	45492
3.1.2.2-	DR MOLTEN ING BGC FI	Υ. Γ	Lb/Hr	31535	14473	383709	20069				2942	ł	398008	850736
TABLE	BALANCE F(SYSTEM US)	1-4	Mols/Hr	1966	7179	13699	456				105	1	23	23428
	MATERIAL POWER	61	Lb/Hr	31535	14473	383709	20069	ł	1	ł	2942	ł	414	453142
		1-2	Mols/Hr	ł	8	ł	ł	1	:	22432	83391	1048	1001	107962
			Lb/Hr	ł	ł	ł	ł	1	ł	717824	2336616	41857	19660	2115957
		n Number	М. М.	16.04	2.016	28.01	44.01	34.08	60.07	32.00	28.02	39.94	18.02	
		Statio	Comp.	CH4	Н2	00	co ₂	Η ₂ S	cos	02	N2	A	H ₂ 0	Total

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(Cont'd)
3.1.2.2-5
TABLE

MATERIAL BALANCE FOR MOLTEN CARBONATE POWER SYSTEM USING BGC FUEL GAS

Stat ion	ı Number	13		14		15-:	18	19-	26	27-	40
.dmo	M.W.	Lb/Hr	Mols/Hr	гр/нг	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr
сн,	16.04	ł	ł	ł	ł	I	ł	ł	ł	1	1
H ₂	2.016	¦	ł	ł	ł	t t	ł	ł	ł	ł	ł
8	28.01	ł	ł	1	ł	ł	l	ł	ł	ł	l
c02	44.01	3011868	68436	2230867	50690	709529	16122	1521338	34568	ł	ł
и ₂ s	34.08	ł	1	ł	!	1	1	}	1	ł	ł
cos	60.07	8	ł	I .	ł	8	1	1	1 1	ł	1
03	32.00	1094976	34218	811040	25345	257952	8061	553088	17284	1	ŧ
N2	28.02	7356091	262530	7356091	262530	2339558	83496	5016421	179030	ł	1
A	39.94	131562	3294	131562	3294	41857	1048	89745	2247	1	ł
H ₂ 0	18.02	1942736	107810	1942736	107810	617870	34288	1324830	73520	903955	50164
Total		13537233	476288	12472296	449669	3966766	143015	8505422	306649	903955	50164

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Referring to Fig. 3.1.2.2-1, fuel gas is received at pressure but must be heated prior to use in the cell. This is done by heat exchange with the anode exhaust gas. When the incoming gas reaches 700F, it is withdrawn from the heat exchanger and sent to a zinc oxide bed for complete sulfur removal. The clean gas is mixed with steam to prevent carbon deposition and then returned to the exchanger where it is heated to 1100F which has been specified as the anode inlet temperature.

In the cell, fuel utilization is 85 percent. That is, 85 percent of the total moles of H_2 + CO are consumed in the cell. For each moles of fuel used, one mole of CO₂ and one of H_2O are released to the gas stream. Thus the mass flow out of the cell is greatly increased over the inlet flow rate. The gas leaves the cell at 1220F and the composition corresponds to the water gas shift equilibrium at that temperature.

As noted above the anode exhaust gas is used to heat the incoming fuel gas. Because of the high mass flow rate of the exhaust gas, it is only cooled to approximately 800F before entering a blower which directs the very low-Btu gas to a catalytic burner. Here the gas is burned and the products diluted with air from the turbocompressor. The resultant gas (Stream 12) is mixed with recycle gas from the anode that has been cooled to maintain proper cell temperature. The mixture is sent to the cathode where the utilization of CO_2 and O_2 is approximately 25 percent. Gas leaves the cathode at 1300F. Part of it is cooled by raising high pressure steam and providing superheat and reheat for the steam cycle. That part is recirculated to the cathode by the recycle compressor. The other part of the stream is expanded through a turbine which drives the compressor and a generator adding to the gross power output.

Heat recovered from the turbine exhaust is used for economizing in the steam bottoming cycle. As mentioned earlier boiling and superheating are accomplished in the cathode recycle stream.

3.1.3 References

- 3.1-1 Handley, L. M.: FCG-1 Power Plant Specifications. Supported by DOE Contract No. EX-76-C-01-2102, March 1980.
- 3.1-2 Benjamin, T. G., E. H. Camara, and L. G. Marianowski: Handbook of Fuel Cell Performance. Supported by DOE Contract No. EC-77-C-03-1545, May 1980.
- 3.1-3 Crawlord, C. D., R. P. Dawkins, and J. R. Joiner: An Economic Comparison of Molten Carbonate Fuel Cells and Gas Turbines in Coal Gasification -Based Power Plants. EPRI AP-1543, September 1980.
- 3.1-4 Interim Project Briefing, NASA-MSFC, Huntsville, Alabama, December 17/18, 1980.

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- 3.1-5 Arthur D Little: Assessment of Fuels for Power Generation by Electric Utility Fuel Cells. EPRI EM-695, 1978.
- 3.1-6 Johnson, W. H.: Study on Phosphoric Acid Fuel Cells Using Coal-Derived Fuels. Ongoing Program Supported by TVA Contract No. TV-53900A.

4.1 Evaluation of Integrated Gasification Fuel Cell Power Systems

The objective of this task is to evaluate the performance of fuel cell systems when integrated with coal gasification and cleanup systems. The gasifiers to be considered are the oxygen-blown Texaco and the BGC slagger. Gasification and gas cleanup systems are the same as reported in Ref. 4.1-1.

Because of their low operating temperature, the phosphoric acid fuel cells do not lend themselves to integration with gasification systems. The more likely use of the acid cell is as a dispersed generator which would have the performance characteristics described in Section 3.1.2. For that reason, this section deals only with gasification systems integrated with the molten carbonate fuel cell.

4.1.1 Identification of Fuel Cell Power Plant with Texaco Gasifier

The fuel cell system with steam bottoming cycle described in Section 2.1.3 will be used as the basis for the integrated power plant. Cell operating conditions are as described in Section 2.1.3 but actual cell output varies with gas compositions and steam cycle arrangement has been changed to fully utilize the heat available in each configuration. A schematic for the Texaco-based system is shown in Fig. 4.1.1-1. A brief description of the system follows.

In the Texaco system coal is mixed with water and the resultant slurry (Stream 5) is pumped up to gasifier pressure. In the gasifier it reacts with oxygen to produce a high temperature (>2300F) medium-Btu gas. At the gasifier exit, a portion of the gas is drawn off to remove the bulk of entrained ash from the main stream. That gas is quenched with water to remove the ash and rejoin the mainstream at the particulate scrubber. Prior to entering the high-pressure boiler, the main gas stream is quenched to below the ash fusion temperature to solidify the ash and minimize heat exchanger fouling. The gas is then cooled by raising high-pressure steam, transferring heat to the clean gas, raising low pressure steam and finally by evaporation in the particulate scrubber. The gas leaves the scrubber fully saturated at 352F. Because it is saturated, there is a large amount of heat available at temperatures below 352F. After reheating the clean gas to 300F and heating the boiler feedwater to 250F, the remaining heat is rejected to ambient and the gas enters the Selexol System at 105F. Following desulfurization, the clean gas is reheated against the raw fuel gas in a series of three heat exchangers. At 700F, the clean gas is removed from the heat exchanger and sent through a zinc oxide bed to guarantee complete sulfur removal. Following this, steam is added to prevent carbon deposition and the gas is reheated to 1100F. At this point the pressure is reduced to the fuel cell operating level by expansion through a turbine.

The remainder of the system is virtually the same as that of the over-thefence case described in Section 3.1. The cell operates at 185 amps per sq ft with an output voltage of 0.81V.

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In the steam cycle, feedwater is sent to the fuel processing equipment where both low- and reheat-pressure steam is raised.

Performance of the system is summarized in Table 4.1.1-1, a heat and mass balance is given in Table 4.1.1-2 and stream compositions in Table 4.1.1-3.

4.1.2 Identification of Fuel Cell Power Plant with BGC Gasifier

The fuel cell system with steam bottoming cycle described in Section 2.1.3 will be used as the basis for the integrated plant. Cell operating conditions are as described in Section 2.1.3 but actual cell output varies with gas composition and steam cycle arrangement has been varied to fully utilize the available waste heat. A schematic for the BGC-based system is shown in Fig. 4.1.2-1.

The major factor in the gasification process that affect the power plant and result in basic differences between the BGC and Texaco based systems are:

- 1. The Texaco slurry feed and high temperature makes it less efficient than the BGC.
- 2. BGC raw gas must be water quenched to remove condensible hydrocarbons and the sensible heat is loss to the power plent.
- 3. BGC product gas contains a significant amount of methane (approx 22 percent of HHV) which cannot be used in the fuel cell.

The primary effect of these differences in terms of the Fig. 4.1.2-1 schematic is the loss of the ability to us: heat from the raw gas to reheat the clean gas leaving the Selexol system. Instead, the heat needed to preheat the fuel gas must be taken from the anode exhaust stream. As a result, the fuel cell part of the system is virtually the same as in the over-the-fence case which has been described earlier.

Because of the gasifier steam requirements, the steam cycle is somewhat different than in the Texaco case. Steam for the gasification process is taken from three sources. Gasifier jacket heat provides approximately 60 percent of the total, 25 percent is raised in the sulfur recovery plant and 15 percent is raised in the sulfur recovery plant and 15 percent extracted from the intermediate pressure turbine. Other steam is extracted from the low pressure turbine to supply process needs at both 50 and 100 paig.

Performance of the system has been summarized in Table 4.1.1-1. A heat and mass balance is given in Table 4.1.2-1 and stream compositions are given in Table 4.1.2-2.

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TABLE 4.1.1-1

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PERFORMANCE SUMMARY FOR INTEGRATED GASIFICATION/ FUEL CELL POWER PLANTS

Gasification System:	Texaco	BGC
Oxident	Oxygen	-+
Coal Rate, TPD	3566.6	
Output:		
Fuel Cell Power, MW	365.3	330.5
Gas Turbine Power, MW	93.2	106.3
Steam Turbine Power, MW	106.0	100.7
Expander Power, MW	35.6	18.0
Gross Output, MW	600.1	555.5
Utilities:		
Selexol Power Consumption, MW	3.3	3.4
Air Compressor Power, MW	45.4	24.8
Oxygen Compressor Power, MW	15.6	7.9
Recirculation Compressor Power, MW	14.8	12.2
Other Utilities Power, MW	16.6	13.3
Total Utilities, MW	95.7	61.6
Total Net Output, MW	504.4	493.9
Heat Rate, Btu/kWh	7223	7362
Overall Efficiency, %	47.2	46.4

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TABLE 4.1.1-2

HEAT AND MASS BALANCE FOR INTEGRATED TEXACO GASIFICATION/FUEL CELL POWER PLANT

<u>Station Number</u>	Temperature (^O F)	Pressure (psia)	Flow (1b/sec) ⁽¹⁾
1	59	14.7	722.0
2	595	150.8	722.0
3	90	17.0	69.4
4	300	734.7	69.4
5			82.6
6	2400	588.0	180.6
7	2400	588.0	162.5
8	2400	588.0	18.1
9	400	564.5	25,5
10	1850	588.0	227.4
11	1200	582.1	227.4
12	742	576.3	227.4
13	522	570.5	227.4
14	404	570.5	227.4
15	403	570.5	252.9
16	352	570.5	259.1
17	352	570.5	64.9
18	360	588.0	64.9
19	352	570.5	194.2
20	334	570.5	183.2
21	199	558.5	154.5
22	105	541.8	152.0
23	70	530.9	126.3
24	300	530.9	126.3
25	700	525.6	126.3
26	665	515.1	220.0
27	1100	510.0	220.0
28	751	155.2	220.0
29	1100	150.0	220.0
30	1220	150.0	541.8
31	1039	145.0	541.8
32	1051	150.8	541.8
33	1113	150.0	1263.8
34	1100	150.0	3936.4
35	1300	150.0	3614.6
36	1300	150.0	942.0

(1) All flow values are for one engine.

TABLE 4.1.1-2 (Cont'd)

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HEAT AND MASS BALANCE FOR INTEGRATED TEXACO GASIFICATION/FUEL CELL POWER PLANT

Station Number	Temperature (°F)	Pressure (psia)	Flow (lb/sec) ⁽¹⁾
37	658	16.3	942.0
38	537	14.7	942.0
39	300	14.7	942.0
40	1300	150.0	2672.6
41	1300	150.0	1959.5
42	1154	150.0	1959.5
43	1300	150.0	713.1
44	1154	150.0	713.1
45	1154	150.0	2672.6
46	1080	145.0	2672.6
47	1094	1:0.0	2672.6
48	108	1.23	267.1
49	108	30.0	257.8
50	108	30.0	9.3
51	338	115.0	9.3
52	250	30.0	257.8
53	255	2836.0	257.8
54	486	2837.0	257.8
55	486	600.0	6.1
56	486	2837.0	251.7
57	592	2400.0	251.7
58	592	2400.0	129.2
59	592	2400.0	122.5
60	662	2400.0	122.5
61	662	2400.0	129.2
62	662	2400.0	251.7
63	1000	2400.0	251.7
64	649	600.0	251.7
65	649	600.0	158.0
66	649	600. 0	93.7
67	634	515.0	93.7
68	486	600.0	6.1
69	644	600.0	164.1
70	1000	540.0	164.1
71	626	115.0	13.0
72	108	1.23	151.1

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TABLE 4.1.1-3

MATERIAL BALANCE FOR INTEGRATED TEXACO GASIFIER/FUEL CELL POWER SYSTEM

Stati	daan Maada	er 1-2		3-6			2	5		2		80	
						Coal	Only						
Comp.	М.И.	Lb/Hr	Mols/Hr	Lb/Hr N	bls/Hr	Lb/Hr	Mols/Hr	Lb/Hr M	bls/Hr	Lb/Hr P	bls/Hr	Lb/Hr M	bls/Hr
ਝੈ	16.04		ł	1	ł	ł	ł	417	26	369	23	48	m
H ₂	2.016	1	1	L 1	ł	ł	ł	18815	9333	16934	8400	1681	656
8	28.01	ł	1	ł	ł	;	ł	380460	13583	342394	12224	38038	1358
83	44.01	ł	ł	•	ł	ł	;	125737	2857	113150	2571	12587	286
H2S	34.08	ł	1	ł	:	ł	ł	11144	327	10020	294	1125	33
SOS	60.07	ł	1	ł	ł	ł	ł	1961	16	841	14	120	7
02	32.00	598880	18715	245472	7671	ł	1	ł	1	1	:	ł	ł
N2	28.02	1949407	69572	2662	95	t a	1	6220	222	5604	200	616	22
×	39.94	34908	874	1797	45	ł	1	1797	45	1598	40	200	Ś
н ₂ 0	18.02	16398	910			297217	:	103831	5762	93452	5186	10380	576
Total		2599593	10006	249931	7811	297217	1	649382	32171	584362	28952	64995	3218

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					T A	BLE 4.1.	1-3 (Cont	(P.					1-9:
		Z	ATERIAL	BALANCE 1	FOR INTECR	ATED TEX	ACO CASIF	IER/FUEL	CELL. POWI	ER SYSTEM			5 534.
Statio	n Number		6	10-	-14	Г	من	16		17-18	~	19	3-9
Comp.	N.N. LI	b/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr M	ols/Hr	Lb/Hr M	ls/Hr	Lb/Ar H	ols/Hr
E	16.04	48	m	513	32	545	34	545	34	144	6	417	26
H2	2.016	1881	933	23216	11516	25099	12450	25099	12450	6284	3117	18815	6333
• 8	28.01	38038	1358	469448	16760	507513	18119	507513	18119	127053	4536	380460	13583
ຮ໌	44.01	12587	286	155135	3525	167722	3811	167722	3811	4199	924	125737	2857
A A S A	34.08	1125	33	13734	403	14859	436	14859	436	3715	109	11144	327
7 SS	60.07	120	8	1201	20	1322	22	13.2	22	300	ŝ	961	16
0	32.00	ŧ		ł	!	ł	ł	1	ł	1	8	ł	1
N2 7	28.C2	616	22	7677	274	8294	296	8294	296	2073	74	6220	222
' <	39.94	200	5	2197	55	2396	60	2396	60	299	15	1797	45
H20	18.02	37337	2072	144574	8023	131912	10095	204221	11333	51123	2837	153098	<u>6496</u>
Totai		61952	2 4714	817695	40608	909662	45323	126166	46561	195490	11656	698649	34905

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R81-	9443	43-9 28-0	r Mols/Hr	1 25	5 2071		1 1336	l 1336 \$ 32562	l 1336 4 32562 	1 1336 4 32562 	t 1336 1 32562	t 1336 t 32562 	t 1336 t 32562 2	1 1336 1 32562 1 32562 2 221 2 25946								
		ĕ	Lb/B1	401	417		37421	37421 1433054	37421	37421 1433054 	37421	37421 1433054 	37421 1433054 	37421 1433054 6192 1757 1757								
	Hal	-29	Mols/Hr	25	9286		13426	13426 1168	13426 116 8 	13426 1168 	13426 1168	13426 1168 	13426 1168 	13426 1168 221 221 44 18731								
	POWER SYSI	26	Lb/нг	401	18721		376092	3/6052 51404	3/6092 51404 	51404 51404 	51404 51404 	3/6092 51404 6192	3/6092 51404 6192 1757	3/6092 51404 6192 1757 3337533								
	EL CELL 1	-25	Mols/Hr	25	9286	13426)	1168	1168	1168	1168	1168 221	1168 221 44	1168 44								
t'd)	IFIER/PUI	23-	Lb/Hr	401	18721	376062		51404	51404	51404	51404	51404 6192	51404 6192 1757	51404 6192 1757								
1-3 (Con	XACO CAS		Mols/Hr	26	9333	13583		2357	2357 327	2357 327 16	2857 327 16 	2357 327 16 	2357 327 16 222 45	2357 327 16 45 45								
ABLE 4.1.	GRATED TE	22	22	22	22	Lb/Hr	417	18815	380460		125737	125737 11144	125737 11144 961	125737 11144 961 	125737 11144 961 	125737 11144 961 6220 1797	125737 11144 961 6220 1797 273					
Ĩ	FOR INTE(1	Mols/Hr	26	9333	13583		2857	2857 327	2857 327 16	2857 327 16 	2857 327 16 	2857 327 16 222 45	2857 327 16 45 <u>554</u>								
	BALANCE	CN.	Lb/Hr	417	18815	380460		125737	125737 11144	125737 11144 961	125737 11144 961 	125737 11144 961 	125737 11144 961 6220 1797	125737 11144 961 6220 1797 1793								
	MATERIAL	0	0	0	0	0	0	0	0		Mols/Kr	26	9333	13583		2857	2857 327	2857 327 16	2857 327 16	2857 327 16 	2857 327 16 222 45	2857 327 16 222 45 45
	-	ir 20	Lb/Hr	417	18815	360460		125737	125737 11144	125737 11144 961	125737 11144 961 	125737 11144 961 6220	125737 11144 961 6220 1797	125737 11144 961 6220 1797 1797								
		on Numbe	м.и.	16.04	2.016	28.01		44.01	44.01 34.08	44.01 34.08 60.07	44.01 34.08 60.07 32.00	44.01 34.08 60.07 32.00 28.02	44.01 34.08 60.07 32.00 28.02 39.94	44.01 34.08 60.07 32.00 28.02 39.94 18.02								
		Stati	Comp.	3	H ₂	8		88	со ₂ Н ₂ 5	co ₂ H ₂ S CMS	α ₂ α ⁰ 2 αs	602 H ¹ 25 COS COS	 ∞2 ∞2 ∞3 ∞2 ∞3 	CO ₂ H ₂ S COS H ₂ O COS A H ₂ O								

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MATERIAL BALANCE FOR INTEGRATED TEXACO GASIFIER/FUEL CELL POWER SYSTEM

		4											34
Statio	n Numb	er	33	34		35		36.	-39	07	-47	48-	3-9
Comp.	м.м.	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr 1	Mols/F	Lb/Hr	Mols/Hr	Lb/Hr M	ols/Hr
CH4	16.04	ł	ł	ł	ł	ł	ł	ł	ł	ł	1	ł	1
н ₂	2.01	و ا	1	ł	ł	ł	1	8	1	ł	ł	ł	ł
8	28.01	ł	1	ł	ł	1	ł	ł	ł	ł	ł	ŧ	ł
со ₂	44.01	1492951	33923	3318178	75396	2468609	56092	643338	14618	1825227	41473	ł	ł
Η2S	34.08	ł	1	1	1	ł	1	8	!	1	ł	1	ł
cos	60.07	. 1	1		1	1	1	ł	ł	ł	ł	8	ł
°0	32.00	542752	16961	1206336	37698	897472	28046	233888	7309	663584	20737	1	ł
N22	28.02	1955600	69793	7503756	267800	7503756	267800	1955572	69792	5548240	198010	ł	ł
A	39.94	36665	918	140669	3522	140669	3522	36665	918	104004	2604	ł	ł
н ₂ 0	18.02	522148	28976	2003464	111180	2003464	111180	522148	28976	1481388	1481388	961781	53373
Total		4550116	150571.	14172403	495556	13013970	466640	3391611	121613	9622443	345032	961781	53373

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TABLE 4.1.2-1

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HEAT AND MASS BALANCE FOR INTEGRATED BGC GASIFICATION/ FUEL CELL POWER PLANT

Station Number	Temperature (^O F)	<u>Pressure (psia)</u>	Flow (lb/sec) ⁽¹⁾
1	59	14.7	860.0
2	595	150.8	860.0
3			82.6
4	282	325.0	154.0
5	163	320.0	134.2
6	105	306.0	132.6
7	70	296.0	122.4
8	1100	291.0	122.4
9	650	281.1	231.6
10	939	276.2	231.6
11	775	152.0	231.6
12	1100	150.0	231.6
13	1220	150.0	527.3
14	1039	148.0	527.3
15	877	146.0	527.3
16	712	144.1	527.3
17	724 ·	150.8	527.3
18	1341	150.0	1387.3
19	1100	150.0	3756.0
20	1300	150.0	3460.3
21	1300	150.0	1091.6
22	655	16.2	1091.6
23	491	14.7	1091.6
24	300	14.7	1091.6
25	250	14.7	1091.6
26	1300	150.0	2368.7
27	1300	150.0	1780.7
28	1126	150.0	1780.7
29	1300	150.0	588.0
30	1126	150.0	588.0
31	1126	150.0	2368.7
32	9 45	145.1	2368 7
33	9 58	150.0	2368.7
34	108	1.23	292.8
35	108	30.0	292.8

(1) All flow values are for one engine.

TABLE 4.1.2-1 (Cont'd)

Station Number	Temperature (°F)	<u>Pressure (psia)</u>	Flow (lb/sec) ⁽¹⁾
36	200	30	292.8
37	250	30	292.8
38	255	2837	292. 7
39	446	2553	29 2.7
40	446	2553	20.6
41	448	415	14.5
42	448	415	6.1
43	6 03	2400	272.2
44	448	415	14.5
45	448	415	6.1
46	448	415	6.1
47	662	2400	272.2
48	1000	2400	272.2
49	648	600	272.2
50	648	600	162,9
51	648	600.0	109.2
52	608	281.1	109.2
53	1000	540.0	162.9
54	1000	540.0	63.6
55	1000	540.0	99. 3
56	108	1.23	99. 3
57	9 37	415.0	4.0
58	510	415.0	24.6
59	611	100.0	39.8
60	475	50.0	4.8
61	108	1.23	15.0

HEAT AND MASS BALANCE FOR INTEGRATED BGC GASIFICATION/ FUEL CELL POWER PLANT

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STREAM COMPOSITIONS FOR INTEGRATED BGC GASIFIER/FUEL CELL POWER SYSTEM

Static	on Number	, _,	-2		e		4		Ś		9
Comp.	MM	1.5/Hr ⁽¹) Mols/H	r Lb/Hr	Mols/Hr	: Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mola/H
CH4	16.04	1	ł	ł	ł	33235	2072	33235	2072	33235	2072
H ₂	2.016	ł	1	ł	ł	14560	7222	14560	7222	14560	7222
80	28.01	ł	-	1	ł	386118	13785	386118	13735	386118	13785
c0 ₂	44.01	ł	ł	ł	ł	20245	460	20245	460	20245	097
H ₂ S	34.08	1	ł	1	ł	11144	327	11144	327	11144	327
COS	60.07	ł	1	1	ł	841	14	841	14	841	14
02	32.00	713312	22291	ł	ł	ł	1	ł			r 1
N2	28.02	2321933	82867	ł	1	8994	321	8994	321		
¥	39.94	41578	1041	ł		160	4	160	4	160	321 k
H ₂ 0 Total	18.02	19534 3096357	1084 107283	297217 297217	1	78675 53972	4366	7154	<u>397</u> 24602	1586 476883	88 24293
(1) A1	l values	based on	one engi	ine.							

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TABLE 4.1.2-2 (Cont'd)

STREAM CONPOSITIONS FOR INTEGRATED BGC GASIFIER/FUEL CELL POWER SYSTEM

	lols/Hr	ł	ł	ł	67934	1	ł	33961	263010	3312	107 940	476146
19	Lb/Hr M	ł	ł	Î	2989291	ł	ł	1086752	7369540	132281	1945079	13522943
	Mols/Hr	1	1	1	33570	ł	ł	16785	82972	1045	34052	163424
18	Lb/Hr	ł	1	ł	1477416	1	ł	537120	2324875	41737	613617	4994765
13-17	fols/Hr	1971	1983	1148	30451	1	ł	ł	105	4	27045	62707
	Lb/Hr N	31615	3998	32155	1340149	t t	ł	ł	2942	160	487351	1898370
12	4ols/Hr	1971	7195	13673	189	ł	ł	ł	105	4	21832	44969
6-1	Lb/Hr ⁽¹⁾	31615	14505	382981	8318	ł	ł	ł	2942	160	393413	833934
7-8	Mols/Hr	1971	7195	13673	189	ł	1	1	105	4	10	23147
	Lb/Hr	31615	14505	382981	8318	ł	ł	١	2942	160	180	475477
Number	м.м.	16.04	2.016	28.01	44.01	34.05	60.07	32.00	28.02	39.94	18.02	
Station	Comp.	CH4	Η ₂	ន	co ₂	н ₂ s	cos	02	N2	¥	Н20	Total

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(1) All values based on one engine.

TABLE 4.1.2-2 (Cont'd)

STREAM COMPOSITIONS FOR INTEGRATED BGC GASIFIER/FUEL CELL POWER SYSTEM

Station	n Number	20	_	21-2	ŝ	26-33		34-61	_
Comp.	.W.M.	Lb/Hr	Mols/Hr	Lb/Hr N	fols/Hr	Lb/Hr ⁽¹⁾	Mols/Hr	Lb/Hr M	ols/Hr
CH_4	16.04	ł	ł	ł	1	ł	1	1	1
н,	2.016	ł	ł	ł	ł	١	ł	ł	1
C 7	28.01	ł	ł	1	1	1	1	ł	ł
c0 ₂	10.44	2208642	50185	696766	15832	1511876	34353	١	١
H2S	34.08	ł	ł	ł	ł	ł	١	١	l
cos	60.07	ł	ł	1	ł	ł	ł	1	١
0,	32.00	802976	25093	253312	1916	549664	12:21	ł	ł
N,	28.02	7369540	263010	2324875	82972	5044721	180040	ł	ł
~ v	39.94	132281	3312	41737	1045	90544	2267	ł	١
0, н	18.02	(945079	107940	613617	34052	1331462	73888	1054224	58503
د Total		12458518	8 449540	393030	7 141817	8528267	307725	1054224	58503

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(1) All values based on one engine.

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4.1.3 Comparison of Performance

The differences between the Texaco and BGC performance shown in Table 4.1.1-1 are due primarily to gasifier efficiency (the slurry feed makes the Texaco less efficient than the BGC), the loss of raw gas sensible heat (in the case of the BGC) and the presence of methane in the BGC gas that cannot be used in the fuel cell but whose energy improves the steam cycle in that system.

Because of the lower oxygen and steam (water) requirements of the BGC gasifier the cold gas efficiency of that unit is much higher than the Texaco units (93% and 77% respectively). However, this difference is offset by the amount of methane in the BGC off-gas. Nearly 24% of the heating value of that gas is attributed to methane and other hydrocarbons that cannot be utilized in the fuel cell. They are, however, used at steam cycle efficiency.

The low exit temperature of the BGC gasifier results in a significant amount of condensibles in the off-gas. As a result, it is not practical to attempt to remove the sensible heat from the stream. Instead, it must be water quenched and the heat is available at less(than 282N.

Referring to Table 4.1.1-1, the effects of these differences can be seen. Fuel cell power is some 10% less in the BGC case. This is a direct result of the difference in the quantity of hydrogen and carbon monoxide in the product gas of each. If only those two gases were considered in calculating cold gas efficiency, it would be 70% for the BGC and 77% for the Texaco gasifier.

Gas turbine power is higher in the BGC system due to the increased flow associated with combustion of methane in the cathode loop. The combustion of methane also raises the steam cycle output. Despite the loss of gasifier raw gas sensible heat in the BGC-based system, steam power is almost the same in both cases. Expander power differs due to the difference in pressure at which each gasifier operates.

While total output of the Texaco system is some 8% higher, net output differs by only 2%. This is due primarily to the air compressor for the oxygen plant and the oxygen compressor. The former is lower in the BGC system since oxygen requirements are 44% lower than for the Texaco gasifier. Oxygen compressor power differs due to the lower flow and also due to the lower pressure to which it must be compressed in the BGC system.

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4.1.4 Off-Design Operation of Gasification/Molten Carbonate Power Plants

The integration of a molten carbonate fuel cell with a gasification system yields a very complex plant. The ability to operate such a plant can only be demonstrated in an actual installation. Simulation of the plant is possible, however, this has not been done to date. Consequently, it is necessary to infer from existing data the probable part-load capability of an integrated gasification/ molten carbonate fuel cell power system.

In Ref. 4.1-2 the plant operation and control of an integrated gasifier/molten carbonate fuel cell was discussed. Two modes of operation at part power were advanced.

The first mode takes advantage of the fact that the powerplant consists of independent gasifier-fuel cell trains operated in parallel. In each train, coal is converted to fuel cell power and the waste heat from all trains is available to a single steam bottoming cycle. Any train can be shut down independently of the others; this reduces the fuel cell output. Shutting down a single gasifierfuel cell train lowers the amount of steam available to the bottoming cycle, thus lowering its output power. This modular unit shutdown approach to turndown has the disadvantage of slow response time to power turnup. It takes two hours or less to bring a hot unit on line. If it has been off line longer than 48 hours and allowed to cool down, it takes a minimum of 12 hours. The second turndown mode has quick response times and infinitely variable power levels. In this mode, all units remain online and power is reduced by changing the inverter controls and reducing the coal flow to the gasifiers.

The total startup time from cold for the powerplant is limited to heat up of the fuel cell and gasifier sections and requires a minimum of 12 hours. A hot restart of the powerplant is expected to take about two hours or less due to the fact that hot components stay near nominal operating temperatures for up to 48 hours following shutdown.

Both complete or partial shutdown of the plant may be accomplished. For partial shutdown to cold condition, the inverters associated with individual fuel cell islands are automatically shut off and disconnected on both the dc and ac sides. Simultaneously, both the coal and air feed to the appropriate gasifiers are stopped causing a halt in the production of synthesis gas. Steam from auxiliary boilers is fed to the gasifiers to control cooling rate. Residual combustible gases in the gasification system and gases formed during the cool down process are burned and vented through the flare stack. After the gasification vessels have cooled sufficiently, they may be depressurized. Residual coal and ash remaining in the vessels is removed through the ash hoppers and carried off through the ash slurry system. Upon removal of the electrical load, the fuel

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cells which are being shut down revert to open circuit conditions. Waste heat is no longer being produced and the cathode recycle loop can be shut down. The turbocompressors are shut off and the cells are allowed to cool slowly by natural convection. If faster cooling of the cells is required, the fuel cell turbocompressor may be operated using its auxiliary burner to provide cooling air to the cells. When the entire plant is to be shut down, the steam bottoming cycle must be removed from the line. Since the steam plant is unfired, it depends on the fuel cell waste heat for process steam production. When the fuel cells are shut down, steam production stops. Residual steam may be either vented to the atmosphere or passed to the turbine condenser unit.

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4.1.5 References

4.1-1 Crawford, C. D., R. P. Dawkins, and J. R. Joiner: An Economic Comparison of Molten Carbonate Fuel Cells and Gas Turbines in Coal Gasification-Based Power Plants. EPRI AP-1543, September 1980. 1

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4.1-2 King, J. M.: Energy Conversion Alternatives Study - Intergrated Coal Gasifier/Molten Carbonate Fuel Cell Power plant Conceptual Design and Implementation Assessment. NASA CR 134955, October 1976.

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1.2 Assessment of Combined Cycle Systems Technology Status

Although the first application of gas turbines in power generating systems occurred in Switzerland in the 1950's, it was not until the late 1940's that utilities in the U.S. began to use the gas turbine. By the end of 1977, there were 47,736 MW of gas turbine capacity installed, approximately 8.6 percent of the total U.S. utility capacity. By the end of 1980, an additional 4700 MW was added.

The vast majority of the installed gas turbine capacity is used in the simplecycle mode as peaking units, i.e., units used less than 1200 hr/yr. In locations where low cost clean fuel is available, simple-cycle units may be used for longer periods, perhaps even as base-load units. There are nearly 5000 MW of combined cycle systems installed or under construction in the United States. These units are usually operated in mid-ranges (1500-3000 hr/yr) although several industrial cogeneration applications are base loaded.

1.2.1 Assessment of Combined Cycle Technology Readiness Development Status

Before projections of future gas turbine system technology can be presented, it is necessary to identify the characteristics of current commercial installations and define the technology readiness and development status of gas turbines.

It is instructive to review, at this point, several basic aspects of gas turbine operation. The simple-cycle gas turbine operates on the Brayton cycle (Fig. 1.2.1-1). As with any thermodynamic cycle, the most important parameter is the peak cycle temperature, i.e., Carnot's law states that the efficiency of a heat engine is a function of the peak cycle temperature and the temperature at which the cycle heat is rejected. In the limit this becomes

$$\eta_t = 1 - \frac{T_r}{T_p}$$

where

n_t = theoretical Carnot efficiency T_r = heat rejection temperature T_n = peak cycle temperature

A second important parameter is the cycle pressure ratio, i.e., the compressor outlet pressure/compressor inlet pressure. There exists a relationship between pressure ratio and turbine inlet temperature which allows selection of the optimum operating characteristics.

In its basic form, a gas turbine consists of compressor, combustor, and turbine components. Typically, the compressor is constructed with one, two or t

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TEMPERATURE-ENTROPY DIAGRAM FOR A BRAYTON CYCLE

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more spools (shafts), depending on overall compression ratio and desired characteristics for engine starting and part-load performance. The combustor is usually designed for compact, high heat release characteristics so that mounting directly on the engine structure is possible. The use of coal-derived gaseous fuels; however, has led to other combustor configurations which will be discussed in later sections. The turbine component produces shaft work, approximately one-half to two-thirds is used to drive the compressor with the remainder available to drive the load, i.e., the electric generator. The electric generator may be coupled to the same shaft as the compressor turbine (single-shaft version) or it may be driven by a frae power turbine (two-shaft version) aerodynamically coupled to the upstream gas generator.

In the combined-cycle mode, a waste heat boiler would be installed in the hot exhaust stream from the gas turbine. Steam raised in this boiler would be used in a steam turbogenerator to produce electrical power. The use of the steam bottoming system reduces the heat rejection temperature from that associated with the gas turbine exhaust to a value associated with the steam condenser. The result is a significant increase in efficiency. The theory of combined cycle operation is treated in Appendix A.2. A more detailed discussion of the steam bottoming cycle is included in the discussions of the system technology readiness.

The following sections discuss currently available and state-of-the-art gas turbines and combined-cycle systems from the viewpoint of their performance and technology readiness. The development trends in turbomachinery components and materials from the current machine through the state of the art to potential future machines are then described.

1.2.1.1 Technology Readiness - Commercially Available Equipment

The present generation of commercially available sim; 'e-cycle gas turbines of interest for utility power generation range in size from approximately 10 MW to nearly 150 MW. These machines have compression ratios ranging from 8.1 to 14.1 and firing temperatures from 1600 F to 2100 F. They are capable of attaining heat rates as low as 10,5~00 Btu/kWh (32.5 percent thermal efficiency). A typical method of presenting performance for a gas turbine is shown in Fig. 1.2.1.1-1 for the FT4C, a nominal 30-MW gas turbine manufactured by United Technologies Corporation.

The actual heat rates incurred by currently installed gas turbines display a wide divergence. This is due to several operating variations, e.g., a machine with many startups and shutdowns will have a high heat rate since the fuel consumption is high compared to the power generated. Also, several utilities use some of their turbines as synchronous condensers, therefore, essentially no power is generat 4 although fuel is being consumed.



PERFORMANCE OF FT4C

1.2-4

The current offerings of combined-cycle systems vary from slightly over 14.5 MW to approximately 675 MW. The lower ratings are for single gas turbine/ steam turbine combinations while the higher ratings are for multiple gas turbines (up to eight) and one steam turbine. The heat rates range from 7500 Btu/kWh to 9500 Btu/kWh (45.5 percent and 35.9 percent, respectively).

The foregoing performance values are based on the use of clean fuels. Use of such fuels is necessary to preserve the lifetime of combustor and turbine (hot section) components. All gas turbine manufacturers agree that natural gas is the best currently available fuel where long life of hot section parts is desired. It is recognized that in most areas of the country natural gas is a limited use fuel and for most utility installations is usually available, at best, only on an interruptable basis. Natural gas costs vary considerably, e.g., $1.00/10^6$ Btu in Texas to $4.50/10^6$ Btu in Illinois.

Next to natural gas, the best fuels for gas turbines are those used in aircraft applications. These are the kerosene fuels, JP4, JP5, and ASTM GT No. 1 also known as Jet A or Jet B fuels. These kerosene-base fuels, although providing good combustion characteristics in gas turbines, are more expensive than natural gas or more plentiful, lower grade fuels. Jet fuel is approximately $$7.00/10^6$ Btu. Below the kerosene-base fuels in acceptability to gas turbine applications are fuels known as diesel No. 2 and home heating No. 2. This latter fuel is widely used by the electric utilities as gas turbine fuel. Residual oil is sometimes used, but requires pretreatment. Even with this treatment, residual oil is not a satisfactory fuel for long-term use. The cost of GT-2 (No. 2 oil) varies from $$1.90/10^6$ Btu in Texas to $$6.50/10^6$ Btu.

Another important aspect to be considered in the design of a gas turbine system is the characteristic of the load to which the system will be subjected. Where the unit is used for short periods at high powers with frequent starts and stops, it is important to have a clear understanding of the number of planned starts and stops in addition to the power level expected during this operation. In addition, it is necessary to know the rate at which the load is to be applied. If it is possible to load the machine at a relatively low rate and still meet requirements, the least deterioration of the equipment over a given operating period will result. If, however, a fast start is required with quick loading, it may be necessary to reduce the length of the overhaul period on the machine.

For a continuously operating installation where loads will be maintained relatively constant, it is desirable to derate the gas turbine to power levels below those needed for peaking applications. Customarily, the base-load or continuous-load power level of an industrial gas turbine installation is approximately 80 percent of the maximum capability of the machine; however, where high ambient temperature are expected year-round, it may be advisable to decrease this to 70 percent or 75 percent in order to stay within desirable turbine temperature levels.

Records compiled from the hundreds of gas turbine installations throughout the U.S. and the world show that the starting reliability of these machines range from 96 percent to 99 percent and overhaul cycle times from 8,000 to 40,000 hours. As might be expected, those engines which have come up to 40,000 hours of operation have normally been found to be lightly loaded.

1.2.1.1 Technology Readiness - State-of-the-Art Gas Turbines

The next generation of gas turbines is already in the testing phase. Generally speaking these turbines have a larger power output and are of heavier duty than the present generation. A joint design by United Technologies and Stal-Laval (Sweden) completed in the mid-1970's, is representative of just such as engine. Known as the UTC FT50 or the Stal Laval GT200, the engine appeared in prototype form in 1975. At the present time only the GT200 is in production although several FT50 prototypes were built and run at the conditions equivalent to 90 MW with a heat rate of 10,000 Btu/kWh. The FT50 and the GT200 have identical gas generators and differ only in that the power turbines operate at 3600 rpm (60 Hz) and 3000 rpm (50 Hz), respectively. This engine could evolve into a 100 MW-plus power plant with heat rates in the 9500 Btu/kWh range.

FT50/GT200

The FT50 is a high performance industrial gas turbine consisting of a twinspool gas generator and a free turbine driving the output shaft. Nominal rotating speeds are 3520 rpm for the low compressor spool, 4200 rpm for the high compressor spool, and 3600 rpm for the power (free) turbine shaft. The two gas generator spool systems are concentric to each other, yet mechanically independent. Each compressor is driven by its own single-stage reaction turbine. The hot gases from the gas generator drive the two-stage free turbine. The free turbine converts the heat and kinetic energy of the gas generator exhaust to useful work energy in the form of shaft horsepower without the use of additional mechanical devices.

The FT50/GT200 was designed as the state-of-the-art gas turbine for use in simple cycle applications such as peaking or mid-range duty. It was designed to handle clean fuels, i.e., natural gas or distillate oil. The engine pressure ratio is 16:1 and it operates at approximately 2100 F with a heat rate of slightly over 10,000 Btu/kWh.

The engine represents the ultimate in adapting aircraft engine technology to industrial applications. The multi-spool design allows each compressor and turbine section to operate at optimum efficiency while the use of the free turbine allows efficient operation at either 50 Hz or 60 Hz.

Because of the multiple shaft design, it is necessary to have a number of bearings, some operating in the hot section. To facilitate engine maintenance, the FT50/GT200 was designed in modular fashion (Fig. 1.2.1.1-2) which would allow the engine to be dismantled, new or rebuilt parts installed and the engine reassembled in ten working days for the longest maintenance item, combustor replacement. This compares to several months for other large industrial machines. -----

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MODULE	WEIGHT (1.8)	TIME TO REPLACE (HOURS)
LOW COMPRESSOR	33,200	12
HIGH COMPRESSOR	20,100	2.3
HIGH TURBINE	4,300	96
LOW TURBINE	5,100	28
POWER TURBINE	33,700	31

ALL MODULE DIMENSIONS ARE LESS THAN 11 FEET

MODULAR CONSTRUCTION OF FT50

1.2-7

In addition to the modular construction, the FT500/GT200 uses aircraft type turbine cooling technologies. (The following section on Development Trends will cover cooling in some detail, therefore, only a brief description is included here.) The technique used in this engine is known as film cooling. Cooling air is circulated within the blade prior to ejection of some portion through holes in the blade leading edge (called showerhead cooling). A film of air is then formed around the blade keeping metal temperatures to the 1500-1600 F range while operating in a 2000 F plus environment. Cooling of this type requires clean fuels since the cooling holes can be plugged by carryover of combustion ash or corrosive components.

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The GT200 has run successfully on the grid of the Swedish Power Grid and the engine is now offered commercially by Stal-Laval. They also offer a combined cycle based upon this design with an output of 728 MW using two GT200 engines and one turbine. The estimated heat rate for this system (on natural gas) is 7420 Btu/kWh. The estimated output and heat rate for the FT50/GT200 are given in Figs. 1.2.1.1-3 and 4.

V84.3

A second, and entirely different state-of-the-art approach to the large industrial gas turbine is represented by another UTC joint effort, this one with KraftWork Union (KWU) of West Germany. In general, European turbine designs have differed from the US designs in that they tended to be less highly stressed both aerodynamically and physically and usually operated at much lower temperatures, i.e., 1350-1600 F, and thus, did not required turbine cooling. These machines generally were designed for base-load (5000 hr/yr or greater) applications.

A typical conservatively designed European machine is the KWU 94 series, a 100-MW plus machine with a single shaft and front-end drive. The exhaust is ducted directly out the rear of the machne making combined-cycle application easier. The method of construction results in a shaft that is extremely stiff and is supported by only two bearings. Rather than the on-line combustion cans used by US manufactures, the KWU combustion system consists of two large externally-mounted silos. These combustors have proven to be extremely durable and are readily adaptable to a variety of fuels, including crude oil, residual and heavy synthetic liquids. The silo construction is also very well suited for low-Btu gas or pressurized fluidized bed cycles, where air must be extracted for the process. The general component arrangement for the KWU gas turbine and auxiliaries can be seen in Fig. 1.2.1.1-5.

United Technologies and KraftWork Union have completed a preliminary design study as part of a potential joint venture, the object of which is to apply advanced UTC technology in compressor, combustor and turbine development to the KWU overall design. The result of such a venture would be an engine design based on the V94 gas turbine but with improved cost per kilowatt, heat rate and emissions

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FIG.1.2.1.1-3

GENERATOR EFFICIENCY = 98.7% ALTITUDE = SEA LEVEL (PAM = 14.7 PSIA)

3600 RPM



ESTIMATED OUTPUT OF FT50

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GENERATOR EFFICIENCY + 98 7% ALTITUDE + SEA LEVEL (PAM + 14.7 PSIA)

3600 rpm



ESTIMATED HEAT RATE OF FT50



performance while retaining the excellent durability that KWU has experienced. Performance estimates of the current V94 and advanced state-of-the-art V94 and V84.3 (60 Hz) are given in Table 1.2.1.1-1.

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Other Engines

At this time, there are no other industrial gas turbines which have reached the state of technology of the FI50/GT200. However, several manufacturers have uprated versions of their current commercial equipment which could now be ordered. For example, General Electric is offering an uprated version of the Frame 7 with a projected output of nearly 79 MW at a pressure ratio of 11.5 and an estimated turbine inlet temperature of 2100 F. Westinghouse presently offers an uprated version of the W501, the W501D, with a rated output of 104 MW at a pressure ratio of 14.2 and a turbine inlet temperature estimated to be 2066 F.

The technology readiness state of gas turbine and combined cycle systems can be described as mature. A number of manufacturers, U.S. and foreign, offer both simple- and combined-cycle power systems burning natural gas or distillate with satisfactory performance. State-of-the-art machines are beginning to appear and could be offered widely on a commercial basis by 1984. These machines could offer an approximate 10 percent increase in size with a five to eight percent reduction in heat rate.

1.2.1.2 Development Trends in Operating Parameters

The following section describes the development trends in gas turbine and combined-cycle technology and briefly discusses the major ongoing R&D programs on turbine technology.

Compressors

Technological advances in aircraft gas turbine compressor design have resulted in substantial improvements in cycle pressure ratios, increasing from about 4:1 in the early English (National Gas Turbine Establishment) centrifugal compressors of the 1940's to present twin-spool axial-flow compressors (see Fig. 1.2.1.2-1). A spool is commonly considered to be a compressor and connected turbine. The compressor or turbine may have one or more stages. Thus, twin-spool compressors consist of two tandem compressors with their respective turbines that form two rotor systems which are mechanically independent but related aerodynamically. These configurations facilitate proper compressor-stage matching for relatively high pressure ratio designs as an alternative solution to single-spool compressors that would have to incorporate variable geometry stator vanes.

The technology is available that has permitted the introduction of machines with pressure ratios exceeding 30:1. Proportionate increases in the number of axial compressor stages to meet these pressure ratio requirements have been

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TABLE 1.2.1.1-1

PERFORMANCE OF LARGE INDUSTRIAL ENGINES

	<u>Commercially Available</u> (KWU)	State of the Art (UTC/KWU)		
Mode 1	V94.2	V94.3	V84.3	
Rotor Speed, RPM	3000	3000	3600	
Output, MW	110	16 1	112	
Heat Rate, Btu/kWh	12000	11140	11140	
First Commercial Appearance	1979	1984	1984	

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COMPRESSOR TECHNOLOGY PROGRESSION

1.2-14

avoided through intensive research and development efforts to increase stage pressure ratios. For example, it is presently possible to achieve single stage pressure ratio of 1.2 to 1.4 while still maintaining stage efficiencies of 90 percent or more. These stage performance levels have been extended to multistage aircraft compressor designs and permit the attainment of average stage pressure ratios on the order of 1.2 to 1.3 with associated polytropic efficiencies of approximately 90 percent or greater. Further improvements in polytropic efficiency to peak values of 90 to 93 percent within the next generation of engines appear feasible. In fact, the LP compressor of the state-of-the-art FT50/GT200 has achieved 90 percent and the HP compressor, 89 percent.

Principal factors which influence stage pressure ratio are rotor tip speed and aerodynamic loading. Rotor tip speeds of 1100 to 1200 ft/sec are attainable with current advanced lightweight compressor designs, and values as high as 1400 ft/sec are forecast for the early 1980's. Although lightweight fan stages are in aircraft operation at tip speeds of 1500 ft/sec and above, high stress limits and supersonic Mach number operation will limit compressors to the aforementioned levels.

A measurement of the aerodynamic loading is the airfoil diffusion factor (D_f) , a parameter which includes a factor reflecting the overall change in relative velocity across the blade row and a term proportional to the conventional lift coefficient. High values of D_f are desirable since they result in a reduction in the number of compressor stages; values of 0.40 and 0.45 are currently obtainable. The present limit in D_f is due primarily to excessive secondary flow or endwall losses and to compressor surge margins that are dangerously low. Further improvements in tip speed and diffusion factor will result from aircraft compressor research and development programs.

The specific flow (1b/sec per unit flow area) of axial aircraft compressors varies from approximately 33 to 40 lb/sec/ft² of frontal area. Increases in this parameter result in more compact designs (smaller cross-sectional area); however, for a given wheel speed (U_m) and work coefficient, high specific flows and, alternatively, higher axial velocities (C_x) reflect high flow coefficients (C_x/U_m) and correspondingly lower compressor efficiency.

Adaptation of these developments to industrial units has been slow because of the lack of incentives to undertake parallel costly development programs for long-life stationary gas turbine power plants. Consequently, present industrial gas turbine axial compressors incorporate moderate blade loadings and tip speeds of only 650 to 750 ft/sec, and achieve average stage pressure ratios only on the order of 1.1 to 1.15. As a result, as many as 15 to 17 compressor stages are required to produce cycle pressure ratios above 10:1. Thus, high thermal efficiency performance cannot be easily achieved with these designs. The FT50 which typifies the advanced technology engines achieves a 16:1 pressure ratio with a 7-stage low compressor and a 10-stage high compressor.

Turbines

Increases in turbine inlet operating temperatures through the use of improved materials and cooling techniques represent the principal means of improving the performance and increasing the output potential of future gas turbine designs. Although historically most of this effort was directed primarily toward aircraft propulsion systems, some of the advances were eventually used to improve industrial designs. This trend is evident by an inspection of Fig. 1.2.1.2-2 which shows the progression of maximum operating cycle temperatures with time for gas turbines designed for various aircraft, industrial, and electric power generation applications. . 1

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The data indicate that turbine inlet temperatures up to 2000 F are currently being used in gas turbine electric power plants, and temperatures several hundred degrees higher are being used in commercial and military a craft. Prior to mid-1960's, advances in materials technology traditionally accounted for a respectable 20 to 40 F increase per year in turbine inlet temperature. From the mid-1960s to early 1970's, however, significant increases in turbine inlet temperature approaching 70 to 80 F per year, were achieved through substantial improvements in turbine cooling techniques in combination with newer materials. Turbine inlet gas temperatures as high as 2200 F for continuous cruise operation and 2600 F for take-off are now experienced by the engines utilized in today's 747 and DC10 jumbo jets.

The projections shown in 116. 1.2.1.2-2 INCLUE that the industrial engines would follow the trends of aircraft engine turbine inlet temperatures, i.e., increase at 50-75 F/yr. However, the rapid increase in fuel costs since 1974 have dramatically slowed progress in industrial engine design. This is simply due to a sharp decrease in turbine sales. Because of the combination of rapid fuel cost increases and confused fuel policies, the utility industry essentially stopped purchasing gas turbines. As sales declined, R&D efforts were reduced by the manufacturers and emphasis was placed on reducing inventories and meeting environmental goals rather than on increasing performance. Thus, turbine inlet temperatures have been essentially constant for several years. The projected increases shown in Fig. 1.2.1.2-2 are based on government or utility industry funded R&D programs aimed at increased performance turbines for use with coal gasification. These programs will be discussed later.

Improvements in turbine performance resulting from technological advances in aircraft turbine designs have paralleled those for the compressor, with turbine isentropic efficiency increasing from about 85 percent in early 1950 designs to over 90 percent in current designs; efficiencies of 92 to 93 percent are feasible in the near future, particularly in high output designs. The design philosophy of turbines for industrial turbine designs will incorporate a higher degree of impulse staging than exists in most present designs to achieve a relatively large gas temperature drop in the nozzle, thus avoiding high gas temperatures with corresponding high cooling flows in the rotor. The stage work for incraft turbines from a low of 20 Btu/lb in the last stage of low pressure ratio


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units to a maximum of about 130 Btu/1b in the high pressure ratio stage of current production units; technology presently exists to increase this value to about 200 Btu/1b without sacrificing turbine efficiency.

High stage work in itself does not cause aerodynamic problems, but when associated with high efficiency it requires high wheel speeds which become limited by structural considerations. Turbine tip speeds in current twin-spool aircraft designs vary from 900 to 1000 ft/sec in the low pressure ratio stages to as high as 1500 to 1600 ft/sec in the high pressure ratio stages. These tip speeds can be achieved in future industrial-type designs that utilize the improved turbine materials and cooling techniques described in the following section.

Problems in Advancing Gas Turbine Technology

The most important future technological challenge to the use of open-cycle gas turbines for generating electric utility power is the need to develop engines capable of economical intermediate- or base-load service on coal-based fuels. Gas turbines operating with steam bottoming cycle offer the greatest potential for attaining this objective.

The most important factor limiting the application of gas turbines to intermediate- and base-load services has been the modest thermal efficiencies attained in comparison with conventinal steam sytems. Efficiency and specific power for simple, open-cycle systems are critically dependent on gas turbine firing temperature (Fig. 1.2.1.2-3). The figure indicates that higher turbine inlet temperatures given higher efficiencies. The specific power term indicates the amount of power which can be obtained from a certain size gas turbine unit and is, indirectly, a guide to cost, i.e., a machine with high specific power will generally have a lower cost in k/kW than a machine of lower specific power.

Further significant improvements in specific power and thermal efficiency for gas turbine power plants can be an achieved only by increasing the turbine inlet temperature accompanied by an increase in the pressure ratio. Any increase in performance due to improvement in aerodynamic efficiencies will be small since these values have reached a relatively high level due to the 30 years of extensive research in compressor and turbines for aircraft propulsion. Thus, the major improvements in performance will result from advances in turbine cooling and materials.

Turbine Cooling Techniques

Only first stage vanes and disks of industrial gas turbines operating at 1600 to 1700 F turbine inlet gas temperatures are cooled. For long life, base-load operation at turbine inlet temperatures of 1800 F and above, successive Į.

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FIG. 1.2.1.2.3



PARAMETRIC PERFORMANCE

stages of turbine blades also require cooling. Cooling is necessary for two reasons, strength and corrosion resistance. With current materials, metal temperatures of 1500 F-1600 F appear to be good compromise between long life and excessive cooling penalty. Turbine cooling can be accomplished with coolants such as air, water, or liquid metals, but because of the complex cooling system designs and mechanical problems associated with liquid systems, air has been used exclusively as the coolant in all aircraft propulsion systems and for all commercially available stationary applications.

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Air Cocling

Turbine cooling systems in current aircraft engines have progressed from single convective cooling configurations incorporating cast, round, radial passages, for vanes (Fig. 1.2.1.2-4) and single cavities for blades to advanced convective heat transfer designs utilizing impingement cooling of the inside surface of the leading edge (Fig. 1.2.1.2-5) to film cooled designs where a layer of coolant is injected in hollow blades through radial slots to form an insulating air blanket for the outside blade surface (Fig. 1.2.1.2-6). These designs were used in the state of the art FT50/GT200.

In transpiration cooled blades, coolant is bled through a porous material which may be formed by a series of small drilled holes along the airfoil surface or a mesh-like material may be used for blade construction. These designs are in the early stages of development and, when applied to 1990 time period turbines, could offer operation at turbine inlet temperatures approaching 2600 F-2800 F.

Another advanced cooling concept currently being demonstrated in test rigs involves the construction of turbine blades and vanes from many thin wafers. A series of radial wafers is fabricated with complex cooling holes (see Fig. 1.2.1.2-7 for a vane and Fig. 1.2.1.2-8 for a blade) glued together to form an aerodynamic shape. The complex cooling passages allow very effective cooling.

Each of the cooling concepts have advantages and drawbacks. For the concepts outlined above, these are as follows:

Radial Holes

- . Low cost
- . Low effectiveness due to difficulty in accurate hole placement
- . Poor low-cycle fatigue life due to poor distribution of coolant

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FIG.1.2.1.2-4



VANE CONVECTIVE COOLING

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CONVECTIVE IMPINGEMENT COOLING

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 FIG.1.2.1.2-6



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FILM COOLING

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RADIAL WAFER COOLING-VANE

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FIG. 1.2.1.2-8

RADIAL WAFER COOLING-BLADE



1.2-25

Convective Multipass

- . Moderate to low cost
- . Moderate effectiveness

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. For high temperature (2000 F plus) applications, showerhead or impingement cooling required for leading edge.

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Impingement

- . Moderate cost
- . Moderate effectiveness
- . Manufacturing problems
- . Tube insert subject to failure

Transpiration

- . High cost
- . High effectiveness
- . Poor structural properties
- . Reduced aerodynamic performance

Wafer Blade

- . Moderate cost
- . Allows design of intricate cooling scheme
- . Subject to cold rib problems
- . High effectiveness

The optimum design will probably be a combination of techniques and fabrication methods.

In aircraft engines, the air used to cool the turbine blades and vanes is at an elevated temperature, 800 to 1200 F, even before it is introduced into the turbine, since it is bled from the compressor discharge airstream. Precooling the compressor bleed air to fairly low levels before it is used to cool the turbine has not been a general practice in these aircraft applications because the added cooling system weight detracts from the potential gains in performance that might otherwise be achieved by reducing the turbine cooling flow. However, as industrial turbine inlet temperatures and compressor pressure ratios continue to rise and approach the limits established for proven turbine blade cooling techniques, precooling of the compressor bled air to 400 F or lower will be considered.

For stationary industrial power plants, air-, water-, or even possibly fuel-cooled heat exchangers could be used to reduce the temperature or compressor discharge cooling air to fairly low levels since in these stationary applications weight is not an important criteria. Therefore, with this technique it should be possible to achieve desirable gas temperature levels on a continuous basis in base-load plants with presently available combinations of impingement-convection cooling techniques.

Water Cooling

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The high effectiveness of water cooling turbine components greatly reduces the cooling air requirements for the gas turbine and offers an opportunity for overall cycle efficiency improvement. If all the turbine airfoils and static structures are water cooled, only a very small percentage of engine airflow will be needed to cool other components such as disks and seals.

There are two types of water cooling to be considered for stationary and rotating components, namely the use of supercritical water and subcooled boiling water as coolants (Fig. 1.2.1.2-9). Cooling by subcooled boiling water will be accomplished in a closed loop. Therefore, the heat transferred in cooling the turbine can be used in the steam portion of the combined cycle. Initially, supercritical water cooling of the blades will not be a closed loop system. There will be a high recirculation rate in the blade and a small amount of steam exiting from the blade trailing edge. However, this will have a negligible performance impact. Further development of the system may produce a closedloop cooling scheme with water recovery; however, the recovered water must be cleaned before being reused as a coolant.

Materials

The improvements in the temperature capability of turbine materials, with time, for aircraft gas turbines is shown in Fig. 1.2.1.2-10. The gains have been most significant for the nicker-base alloys, with an improvement of more than 200 F in temperature capability over the past decade. Presently, nickerbase cast blade alloys, such as Inconel 718, B1900, and IN-100, have in some aircraft turbine designs replaced forged Udimet 700 (U-700) for high-temperature applications and, at the same time, decreased thermal fatigue failures by a factor of five.

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WATER COOLING

 FIG.1.2.1.2.10



AIRCRAFT MATERIALS TEMPERATURE IMPROVEMENTS

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STRESS-TEMPERATURE RELATIONSHIP

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Figure 1.2.1.2-11 shows the stress-to-rupture strength of the nickel-base alloy IN-100 for a range of material temperatures. Curves for 1000-, 10,000-, and 100,000-hr lifetimes are shown and illustrate the sharp reduction in allowable stress as the desired operating time is increased. The 1000-hr life curve is based on experimental data while the 10,000- and 100,000-hr curves are extrapolated data from short duration test results. Although these materials were developed initially for relatively short time (1000 hr) high-strength aircraft turbine applications, modifications to the heat treatment cycle have permitted their use for long time operation (approaching 100,000 hr) as required in baseload power generation. It is anticipated that similar modifications for the aforementioned high-temperature nickel-base alloys will allow the eventual replacement of such alloys as wrought Udimet 5000 and M252 presently used in industrial turbines and, thus, permit high operating temperature.

Turbine blade materials for industrial designs anticipated by the 1990's will be the superalloys currently under development for advanced aircraft gas turbines; these include unidirectionally solidified eutectic alloys such as Nk_3Al-Ni_3Ct and particle or fiber dispersion-strengthened metals. On the basis of efforts currently underway it is reasonable to expect a continuation of at least 20 F-per-year improvement in material temperature. Chromium-base alloys, for instance, with potential firing temperatures to 2000 F, are being investigated. While recent breakthroughs in the ability to coat columbium-base alloys also offer a possibility for the use of these alloys, it is currently felt that advanced cooling schemes could, however, allow relatively low metal temperatures which, in turn, could allow the use of current, less expensive materials.

Ceramic materials such as silicon nitride and silicon carbide composites are under intensive study for automotive gas turbine applications and offer firing temperatures on the order of 2500 to 2600 F. The quality and reliability of ceramic materials have been greatly improved during the past decade. The benefits of ceramics include intrinsically low cost, abundant material, low density, corrosion resistance, and high temperature strength. Both silicon nitride and silicon carbide are being studied. The use of ceramics in gas turbines can be expected when the industry's design systems can accommodate materials that exhibit low ductility, large thermal expansion differences, and when stress concentiations can be eliminated by design and subsequent fabrication.

Coatings

Coatings have been developed which provide adequate oxidation-corrosion resistance for blades and vanes in aircraft turbines. These coatings are aluminide types, applied by pack or slurry techniques. These coatings have eliminated intergranular oxidation attack (which could cause a thermal fatigue failure) and have provided some protection against sulfidation attack (corrosion

due to the presence of sodium sulfate compounds in the combustion products of distillate fuels). However, coating life decreases rapidly with increases in allowable metal temperatures. For instance, at a metal temperature of approximately 1500 F, a coating life of 50,000 hr may be achieved. A 200 F increase in metal temperature reduces coating life to less than 10,000 hr. Consequently, coating life rather than creep strength properties as determined by peak temperatures in the first-stage turbine vanes will be an important criterion used to specify turbine inlet gas temperature for different power plant operating mode. Adequate coating life could be especially critical when the fuel used in the engine contains quantities of ash, metals, or sulfur that could cause erosion and corrosion problems. Natural gas or low heating value coal gas, in this respect, would be ideal fuel for high performance gas turbines. Meanwhile, new coating materials and application techniques including nondiffusion-type coatings are being developed which, when applied in sufficient thicknesses, should provide uninterrupted service for period exceeding 25000 hr.

Ceramics

Ceramics will offer the advantage of large reductions in cooling heat load throughout the turbine system as reliable methods in incorporating brittle materials are developed. Early application would include stationary structures such as vanes, platforms and outer air seals. Because of the requirement in industrial machinery for long life, low cost systems, maximum metal temperatures must be kept low. While development of advanced cooling schemes will make this possible, additional advantage can result to the cycle if the heat loss could be avoided in part by the use of materials with higher surface temperature capability. Ceramics of various types offer this potential.

The fundamental problem to solve in realizing this potential is adapting design procedures to an unforgiving brittle material. Two approaches have been followed in recent years to use ceramics in gas turbines: (1) the use of insulating graded ceramic coatings bonded to metal members which carry the structural loads, and (2) the use of bulk ceramics as high-temperature structural members. The problem in the first approach is to develop bonding methods and cooling systems which ensure that large differential thermal strains do not occur throughout the operating range. These strains could impose excessive stresses on the ceramic or its bond, resulting in spalling of the coating or crack propagation into the substrate metal. The successful application of such ceramic coatings have used sprayed or sintered powder ceramics bonded to the metal substrate chrough a ceramic/metal intermediate layer which is graded form pure ceramic at the interface with the ceramic surface layer to pure metal at the substrate interface.

While bulk ceramics potentially offer minimum performance penalty and low cost, the problems to be overcome are: (1) obtaining uniformity of material properties, and (2) developing design technology to provide transmittal of

loading throughout the ceramic and from the ceramic-to-metal structure without excessive stresses, even locally. A fundamental of brittle material design is that failure initiated at a point of weakness or stress concentration is not absorbed by ductile flow but leads to a fracture. Considerable progress has been made toward acquiring the required materials strengths and design technology, however, the appearance of bulk ceramics is expected to be several years after that of graded ceramics.

Combustors

All of the currently offered large industrial gas turbines from U.S. manufacturers have the in-line, can-type combustor (see Fig. 1.2.1.2-12). This combustor has a long history of development in aircraft use and has many advantages: high volumetric heat release, ease of cooling, and low cost, for example. These advantages, however, are dependent on burning clean, high calorific value fuels such as natural gas or distillate. Several European manufacturers such as Stal-Lava?, Brown Boveri and KWU use off-base combustors. They are usually tall cylindrically shaped devices sometimes called silos (Fig. 1.2.1.2-13). The requirement to burn coal-derived fuels, both liquid and gaseous, and to meet new stringent environmental regulations (discussed in Task 2) has led to development of alternative burner configurations.

The major problem to be faced is that of reducing the emissions of nitrogen oxides (NO_x) . There are two types of NO_x emissions, thermal NO_x due to the fixation of atmospheric nitrogen during high-temperature combustion; and fuelbound nitrogen caused NO_x due to combustion of fue! containing nitrogen bearing compounds. Unfortunately, there is not a simple way to handle both problems.

Thermal NO_x production is a strict function of combustion temperature (Fig. 1.2.1.2-14). By modifying the combustion process to reduce peak temperature, thermal NO_x can be reduced. One method of doing this is to premix fuel and air to the desired stoichiometric ratio prior to combustion, i.e., burn lean ($\emptyset = 0.6-0.7$ for example). This type of combustor modification affects only the combustor "head" and can be used with any type of combustor. A significant amount of testing on the fuel gas from a Texaco gasifier has been carried out by UTC using a modified can-type burner using a premix combutor head (Fig. 1.2.1.2-15). This premixed type burner has shown that it can easily meet NO_x standards using coal-derived fuel gases.

To reduce fuel-bound nitrogen caused emissions, a different type of modification is required. By burning rich (\emptyset >1), the formation of NO_x from fuel-bound nitrogen is suppressed. The resulting partially burned mixture is then quickly cooled by mixing with excess air and the combustion completed at a reduced temperature to ensure low thermal NO_x emissions. Testing of this rich-burn/quick quench burner (shown schematically in Fig. 1.2.1.2-16), again by UTC, has shown great promise.

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FIG. 1.2.1.2.12

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TYPICAL IN-LINE COMBUSTOR

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CROSS-SECTION OF SILO COMBUSTOR

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EQUIVALENCE RATIO EFFECT ON NOX EMISSIONS

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PREMIX COMBUSTOR HEAD



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SCHEMATIC OF RICHEURN QUICK-QUENCH COMBUSTOR

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The major development trend in combustor technology is, then, the low pollution burner. A second consideration is the reduction of maintenance and increased component lifetime. The use of the off-base, silo-type burner appears to be a method of attaining this goal. The large size means that the volumetric heat release is reduced, thereby reducing the cooling necessary to maintain reasonable combustor wall temperature. The use of ceramic liners, usually segmented into tiles or other smaller shapes, further reduces burner wall cooling requirements. The large volume also ensures complete combustion of low heating valve fuels. The shape is also conducive to two-stage burning, i.e., the rich burn/quick quench approach. The major disadvantage is the increased manufacturing and erection costs.

Steam System Development

To this point, the steam system technology has been essentially ignored. This is because the steam systems used with combined cycles have generally lower operating parameters than the large central station systems. For example, nearly all of the fossil-fired steam stations installed in the past two decades operate at 1000 F and 2400 psi or higher throttle conditions with reheat to 1000 F. There is no single combined-cycle operating level although most are limited by gas turbine exhaust temperatures to throttle conditions of 800-900 F with a corresponding pressure of 1450 psi or less.

As turbine inlet temperatures increase, there is a corresponding increase in exhaust temperature and, thus, steam conditions will increase. One way of indicating the trend in steam cycle equipment is to examine the relationship between turbine exhaust temperature and amount of heat transferred to the steam system. This is usually done through the use of temperature-heat or T-Q curves. In Fig. 1.2.1.2-17, a T-Q curve for a combined cycle system using a typical currently available bottoming steam system is shown. A minimum pinch (smallest temperature difference between the gas side and the steam side) of 50 F is usually deemed reasonable and a maximum stack temperature of 300 F is desirable. While this type of combined cycle is attractive, system efficiency can be increased by better utilizing the turbine exhaust heat. For example, a dual-pressure system (Fig 1.2.1.2-18) which has a high-pressure and a low-pressure boiler section allows more steam turbine power to be generated while meeting the various pinch temperature constraints.

As turbine inlet temperatures increase, and the turbine exhaust temperatures become higher, it is possible to use steam cycle technology comparable to that used in central stations. In Fig. 1.2.1.2-19, a 2400 psi, 1000 F/1000 F steam system is shown. This would bottom a gas turbine having a turbine inlet temperature over 2400 F (the exact value is a function of engine pressure ratio). 1.000

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1450 psia/900F

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COMBINED-CYCLF WASTE-HEAT BOILER T-Q CURVE FOR NONREHEAT STEAM CYCLE

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FRACTION OF TOTAL HEAT TRANSFERRED - Q

COMBINED-CYCLE WASTE-HEAT BOILER T-Q CURVE FOR DUAL-PRESSURE STEAM CYCLE

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COMBINED-CYCLE WASTE-HEAT BOILER T-Q CURVE FOR REHEAT STEAM CYCLE

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The trend in steam equipment is such that no new technology need be developed. However, variations on waste heat boiler design will be necessary.

Current Research and Development Efforts

Several research and development programs are currently underway on turbine cooling and materials for advanced gas turbines. The two major sponsoring agencies are the U.S. Department of Energy (DOE), and the Electric Power Research Institute (EPRI), the central research organization of the U.S. electric utility industry.

DOE Programs

The major DOE-sponsored program currently underway is the "High-Temperature Turbine Technology Program." This program has as the objective, the development and demonstration of the technology to allow operation at a turbine inlet temperature of 2600 F on a coal-derived fuel, specifically low calorific value gas. Although the technology demonstration does not have to be at full scale, all technology must be applicable to machines of a nominal 100-MW capacity.

This program has three phases. Phase I, which commenced in June 1976, defined an overall power plant concept and identified the approach to attaining the 2600 F goal. In Phase II, a two-year program started in September 1977, testing of components and cooling schemes in rig scale units was to be carried out. The third and final phase, which was originally scheduled to last from September 1979 to September 1982, would result in the demonstration of a turbine hot section (combustor, diffusers, vanes and blades) operating on gasified coal. The third phase has not yet been sent to procurement.

The four major U.S. gas turbine manufacturers, Curtis Wright, General Electric, United Technologies and Westinghouse participated in Phase I. Their proposed schemes are as follows:

Curtis Wright - transpiration cooling of both vanes and blades by air General Electric - water cooling of both vanes and blades United Technologies - water cooling of static structures and vanes with air cooled blades (film cooling)

Westinghouse (with the assistance of the Allison Division of General Motors) - transpiration and film cooling by air of vanes and blades

Only the first two manufacturers were refunded in Phase II; a new competition will be held for Phase III contractors. The results to date in these programs will be discussed in Section 1.2.2.2, Demonstration Engine Characteristics.

A second program currently being carried out under DOE sponsorship which has applications to advanced gas turbines is the so-called "Fireside Corrosion" program being performed by General Electric and Babcock & Wilcox. The objective of this program is to identify the corrosion characteristics of a variety of materials suitable for use in gas turbines and in boilers when exposed to the combustion products of coal or coal-derived fuels.

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Under DOE sponsorship, the United Technologies Corporation developed a technology for a low-cost, high performance gas turbine for peaking services. This machine of nominal 10-MW capacity uses a combined axial/centrifugal compressor, patented diffuser concepts and a "wafer" blade turbine operating in the 2200-2400 F range.

Specific goals for this program were to demonstrate the turbine cooling and high-pressure compressor technologies required to achieve thermal efficiencies of 34 to 36 percent simple cycle while reducing the number of compressor and turbine parts 80 percent over state-of-the-art units. Heat transfer cascade testing of the first-stage turbine wafer airfoil cooling designs has exceeded design goals (1500 F maximum metal temperatures at 2500 F combustor exit temperature) and demonstrated cooling effectiveness levels higher than current production aircraft engine airfoils. Additionally, the all-convective cooling design eliminates the leading edge showerhead holes which would be susceptable to plugging in an industrial fuel environment.

The advanced single-spool compressor employs low aspect ratio axial compressor blading and a centrifugal back end stage to achieve 18:1 pressure ratio at 90 percent polytropic efficiency in only 7 stages. The compressor rig tests results were expected during the last half of 1980.

ARPA Programs

The major program on advanced non-metallic materials for gas turbines was that carried out under the sponsorship of the Advanced Projects Research Agency (ARPA) of the U.S. Government. A multiyear program was carried out by Westinghouse and the Ford Motor Company on the development of ceramic turbine components. Westinghouse was to develop ceramic stators for large industrial engines while Ford was to develop ceramic components for automatic gas turbine applications.

Although Westinghouse developed the technology to fabricate large ceramic vanes, successful long term application has not been demonstrated. The major difficulty appears to be the dissimilar thermal growth of the ceramic vane end of the metallic mounts holding the vanes. Failure occurs after only a few cycles.

Ford, however, appears to have developed the technology required to fabricate smaller ceramic components and utilize them successfully in the several hundred horsepower-size engines.

Currently, Gairett Air Research is continuing this program with the objective of developing ceramic technology for industrial turbines of 2 to 10 MW.

EPRI Programs

Additional research efforts have also been carried out under the sponsorship of EPRI. A multiyear program beginning in 1975 on water cooling of static and rotating turbine parts has been performed by General Electric. This program has as its goal, the development of gas turbines able to operate at 2800 to 3000 F on coal-derived liquid fuels. A small rotating test rotor with blades operating at 2800 F has been demonstrated by General Electric during this program.

A test program on erosion/corrosion of turbine materials in coal combustion products and particulate-laden streams is being carried out by United Technologies under EPRI funding.

The latest research program supported by EPRI is the "High Reliability Gas Turbine Combined Cycle Development" program. Recognizing the benefits offered by the gas turbine for future utility power plants, EPRI has asked the gas turbine manufacturers to create a new engine design especially tailored to fill that role. It is envisioned that the gas turbine can be used advancageously in combined-cycle power plan's to satisfy base/intermediate-load generating requirements. This represents a shift in emphasis from the current use of gas turbines principally for peakers; and, it presents a more dema ding set of requirements. The gas turbine of the next generation must be designed primarily to operate dependably and reliably. The new design must do this while fulfilling the overall need to produce electric power at a low cost and exhibit increased reliability over the current market offering as a proeminent requirement. This is to be achieved with minimum use of hardware development process. Rather, the new high reliability machine must represent the application of perfected and credible analytical design procedures. Westinghouse and UTC was funded by EPRI for this program. At this time, no release has been made of Westinghouse results.

The program has been mated with the joint venture design study that is being considered by UTC and the Kraftwork Union Company of West Germany. This new 60cycle engine design, designated the V84.3 is an improved version of the KWU production model V94.2. The changes to the compressor section include an optional variable inlet guide vane for improved combined-cycle part-load performance. The first four stages were redesigned for a 10 percent increase in airflow and a 20 percent increase in pressure ratio. The turbine hot section was also redesigned by increasing the turbine diameter, increasing the firing temperature and reducing the airfoil metal temperatures. These changes significantly increased the turbine output while improving the hot section durability. This program was funded through the end of 1980 with most of the effort devoted to the evaluation of the UTC/KWU engine design.

1.2.2 Develop Data Base on Commercial and Demonstration/Pilot Systems

A data base on commercially available gas turbines and combined-cycle systems is presented for the major U.S. and foreign turbine manufacturers. A second set of data is presented for state-of-the-art systems that are proposed for operation within the next several years.

1.2.2.1 Commercially Available Systems

Data on commercially available gas turbine and combined cycles systems are given in Tables 1.2.2.1-1 and 1.2.2.1-2, respectively. Only turbines of 10,000 kW or larger are listed. Machinery from Eastern-bloc countries have been omitted.

1.2.2.2 Demonstration/Pilot Systems

Projections of operating characteristics of advanced gas turbines which could result from various DOE- and EPRI-sponsored programs are given in Table 1.2.2.2-1. Also included are the characteristics of an advanced Westinghouse turbine which is to be offered commercially in 1981-82. 1.2.3 Evaluate Requirements for Operational System/Subsystem Development and Testing

The technology required for the advanced gas turbine is already under development. Both DOE- and EPRI-sponsored efforts previously discussed will provide component technologies which could be used in advanced turbines. A brief description of the status of these programs follows.

1.2.3.1 High Temperature Turbine Technology Program

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The High Temperature Turbine Technology Program (HTTTP) is a three-phased program initiated in 1976 to develop to a technology readiness state the requisite hot section, i.e., combustor or turbine, components needed to operate at 2600 F and above. The three major U.S. turbine manufacturers plus Curtis Wright were awarded Phase I contracts. In the fall of 1977, Curtis Wright and General Electric were selected to carry out Phase II development and testing.

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U. S. Manufacturers

	Year	Base	Load	Peak	Load				Temperat	". "
Mode I	First Appearance	Output	Heat Rate Btu/kWh	Output	Heat Rate Btu/kWh	Pressure Ratio	Flow Rate 1b/sec	Speed	Turbine Inlet	Exhaust
Curtiss-Wrigh	t Power Systems									
Mod-Pod 20HEL	1973	20,000 kW	9,800	20,000 kW	9,795	18.0	144	3600/3000	2150 est.	888
Mod-Pod 25	1974	24,900 kW	11,850	26,00 kW	11,770	11.0	236	3600/3000	1990 est.	566
Mod-Pod 50	1974	49,800 kW	11,850	53,400 kW	11,770	11.0	472	3600/3000	1990 est.	566
General Electi	ric GT Division									
63142	1949	10,200 kW	13,540			7.2	113	6,500	1730	619
G3132R	1951	9,750 ku	10,520	1	-	7.2	113	6,500	1750	668
G5261	1958	18,900 kW	13,400	1		8.1	213	5,100	1720	955
C5341	1958	24,700 ku	12,230	1		10.2	261	5,105	1730	106
PG5341	1961	24,110 ku	12,450	25,050 kW	12,340	10.2	261	5,105	1805	955
67851	1791	61,750 kW	10,920	-	-	9.7	529	3,600	1850	544
PG7851	1970	60,000 kW	11,130	66,300 ku	11,030	9.7	527	3,600	1960	1018
G7101	1977	75,000 ku	10,590	1		11.5	290	3,600	1985	116
PG7101	1976	72,900 ku	10,790	78,800 kW	10,750	11.5	587	3,600	2085	1045
C9141	1976	105,600 ku	10.700	-		9.6	160	3,000	1955	953
PC9141	1976	102,700 ku	10,900	111,300 kW	10,860	9.6	753	3,000	2055	1020
LM2500	1979	20,100 kW	9,800			18.0	144	3,600	2167	076
PG6501	1978	34,850 ku	11,310	38,000 kW	11,190	11.5	302	5,105	2085	1055
66501	1978	35,650 kW	11,170		1	11.5	302	5,105	1985	686
UTC Pover Sys	tems									
FT4C-3F	1974	29,820 ku	10,880	32,180 kW	10,760	13.8	306	3.600	1970 est.	877
TP4-2(C3F)	1974	59,950 ku	10,820	64,700 kW	10,700	13.8	611	3,600	1970 est.	877
Westinghouse	Combustion Turbin	e Systems Divisi	uo							
W2518 60 Hz	1974	38,850 kW	11,980	41,960 kW	11,845	12.0	363	5,015	2066 est.	1111
W2518 50 Hz	1974	38,850 ku	11,980	41,960 kW	11,845	12.0	363	5,011	2066 est.	1111
W501D 60 Hz	1975	96,440 kW	10,660	104,200 kW	10,580	14.2	173	3,600	2066	1026
W501D 50 Hz	1981	95,250 ku	10,800	102,880 kW	10,720	14.2	794	3,000	2066	1026

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Come			Vestinghous			General Elec																		Nolls Toyce Av	Bolle Royce Sp	Bolls Proce BB	Bolls Rove 01	Canada I Flanne	generator	General Electr		[-sugar [0 out		
4.ean	Exhaust		9446		IVe	505	166	576	056	856			1013	1022			186	1000	£		990		616	176	1	1	500	1	542	181	783	968	968	
Temperat	Inlet		1440 2070		1710	1730	1985	1840	0741	1955			1850(est)	1950(est)		1 TENCO	1400(est)	[750(est)	(issinger		1691	1652	1787	1827	1	1	0661		2160(est)		1	1735	56.1	
	Speed		5,000		100	5,100	5,705	3,000	3,000	3,000			\$8773/4782	3,600			4.850	4.920	-		3,000	6.500	6.500	6,500	6.500	6.500	3000/3600		0.500	3000/3600	3000/3600	3000/3400	2000/ JP00	
	1b/sec		999		240	266	300	160	752	898			355	619		71.1	192	355	1		192	168	168	170	I	1	242		141	212	244	184	1	
	Fatio		10.0		10.0	10.0	6.11	-	1	ú			8.8	0.11			7.0	11.0			9.6	9.5	9.8	9.5 2.9	1	20.02	0.11		17.2	30.0	0.04	10.9		
beo	Btu/Auf		11,267		-	12,310	11.170	-	10,930	10.790			11.890	10.749		14.520	12.975	11,310			11,860	12, 390	12,390	12,940	10,410	10.270	11.020	010 0	9.424	9,280	10714	12,300		
Peak L	Output		41.510 84			25,500 kg	10, 601 22		93, 900 ku	113, 300 kg			37,500 84	78.900 kV		8.950 M	19.500 84	41,180 84			B0,600 ku	14.580 kg	15.960 84	17, 550 84	13, 230 84	22,600 10	55, 800 11	20.650 84	21.890 44	36,090 ku		82.000 ku		
peo	Btu/kin		11,564		12.020	12,310	\$11°11	10,820	11,050	10,820			12.015	10,756		14.840	13,175	11,590		110 000	12,030	13,240	13,040	12, 780	10,800	19.480	11, 300	126.6	9.452	9.637		13.260		
Base 1	Output		38.160 ku 92.980 ku		25, 350 84	24, 500 ku	- m m. Cf	87,400 14	84, 700 km	104 . 500 km			34, 500 kg	72, 500 44		8,200 kW	18.200 14	92,400 ku		4 750 ku	74.000 24	11, 100 84	12, /80 14	14, 190 14	11.200 64	10 DCL 02	46, 700 ku	13,820 64	20.060 km	31,000 kM	20 000 10	58, 800 km		
Tear	Appearance		1979 1979	5	1958	1991	11/1	1976	1978	1978	1 + Co Ltd	bounchinery inc	1970	1970		1964	8961	1412		1940	5161	1965	5/61	6261	0961	2101	2161	6161	1978	1979	0401	5161		
	441	FC	1011	schon-At Lant 14	141	5341		111	141	1916	C. Brown, Bover	The Bovert Tur			11 TIC			9	Gas Turbines	-	014	661-1	-1167	2/511-5	1-1	10	+10	-125/2	-125/6	051-8	~	-2		

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IABLE 1.2.2.1-1 OPERATING CMARCTERSTICS OF COMPERCIALLY AVAILABLE CAS THERINES (COML.)

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L-955	343	-9			Ironsee															M2500 gas generator	M5000 gas generator				icenses										
	Comment				General Electric 1												Stal-Laval G735			General Electric L	General Electric I.				General Electric L										
	Ture. 'F	Exhaust		-	756	1	1016	1038	953	850	196	1023	1045	1020	1020		750		942	986 (824	968 (616			967		1045	1	922			985	
	Tempera	Turbine Inlet		-	1805	1	1950	2085	1950	IRUS	1950	1950	2085	1950	2055		1560		1700 (est	2140 (est	2120 (est	1/00 (est		1730	1805	-	1950		2085		2055			1800	0001
In mont call		Speed		6 500	5.105	5,105	3,600	3,600	3,000	\$ 105	\$105	3,600	3,600	3,000	3,000		3,000/3,600		3,000/3,600	3,000/3,600	3,000/3,600	6,400		6,500	•,105 •,105	001 5	5.100	009-1	3,600	3,000	3,000			3,000	
		Flow Rate Ib/sec		116	261	304	530	165	760	358	101	524	665	153	153		185		161	149	289	191		115	266	101	301	909	599	876	868			763	
		Pressure Ratio		1.1	10.0	0.11	9.7	11.5	9.6	10.2	11.0	9.7	11.5	9.6	9.6		12.0		12.6	18.1	28.8	5.1		7.1	10.1	5 11	11.5	5.11	11.5	11.4	11.4			0.01	
Manufacturers	peor	Heat Rate Btu/kUh		-	12.180		10,860	10,660	10,870	17.120	11.310	11,030	10,750	11,090	10,860		11,780		13,760	9,550	9,080	1		-	12,220		11,210		10,640	-	10,790			11,730	702.00
Foreign	Peak	Output			15.950 ku		66, 300 kW	80,300 kW	MX 005' 76	25, 200, 64	NA 000. M	64.600 ku	78.800 ku	91,600 ku	111,300 ku		14.000 kW		11,000 ku	21,800 kW	38,100 ku	1			26.450 ku		34,750 kW		80,500 ku	1	113,300 kW			81,300 ku	
	peor	Heat Rate Btu/kWh		13 540	12,310	11,250	10,960	10,700	10,790	17 440	11.380	11,130	10,790	11,210	10,900		11,960		14,000	9,580	9.210	14,200		13,320	12,020	11 150	11,280	10.510	10.690	10,620	10,830			11,790	
	Base	Output		10.200 ku	24.050 kW	31,800 ku	60,000 kW	74.290 ku	85,200 kW	14 057 16	31.100 60	58, 500 ku	72,900 kW	82,600 kW	102,700 ku		12,300 kW	ustries	10,000 ku	20,000 ku	35,400 ku	10,050		10,450 ku	25, 350 kW	11 027 61	31,750 kW	76.400 ku	74.400 ku	107,500 ku	21, 100 ku			74.460 ku	
	Year	First Appearance		1966	1967	1991	1972	1981	1981	1967	1981	1972	1981	1980	1961		1979	ar Ima Heavy Indu	1968	1976	1978	9/61	Incering	1949	1961	1978	1978	19.6	1976	1979	1979	5		1971	1
		lode1	Htachi	67115	1465	15441	27821	18610	11162	17120	C6441	PG7821	1867981	111634	171604	Htachi Zosen	5615	shikava]ima-Ha	M1500	M2500	1M5000	Lype /	John Brown Eng	53142.1	65341P	A4411A	PG6441A	21015	C7101E	29141E	PC9141E	trafe Morb Date	THAT WILL BOLK AND	2.693.2	

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		-	1.6.6.8 B	Foreign	Manufacturer						-9
									Temperatur	40 · .	
	Year	Base	Load Heat Rate	Peak	Heat Rate	Pressure	Flow Rate	Speed	Turbine		Comments
Mode 1	Appearance	Output	Btu/kWh	Output	Btu/kMh	Ratio	1b/sec	ed.	Inlet	Exhaust	
Kvaerner Br	To.										
1111	1958	24. 700 44	12.230	26,500 ku		10.2	266	5,105	1805	556	General Electric License
146534	1961	24.000 14	12,460	25,900 ku	12,350	10.2	266	5,105	1805	556	
C6001	6161	31,800 ku	11,250	33,300 kW	-	11.5	300	5,100	1950	1961	
69001	1979	105,600 ku	10,700	114,000 kW	10,860	9.6	868	000.0	5507	0701	
		103 200 501	10 900	11.300 kv	10.860	9.6	868	3,000	2055	1020	
100634	1978	20.100 60	9.800	1	1	18.0	144	3,600	1	1	
1142500	8261	19, 730 ku	066.6		1	18.0	144	3,000	I	1	
Mitsubishi	Heavy Industries										
							571	4 000	1450(est)	909	Westinghouse License
MN-101L	1968	8,710 kW	14,960	NA 067.6	0.0.0	0.0	276	6 012	1475(est)	784	
191-WM	1963	17.790 ku	13,440	19,100 kM	13,220	0.1	607	4 984	2050(est)	1008	
Mu-251	1964	36,840 ku	003,11	39,980 ku	11,520	0.21	140		In the second		
MU-5018	1073	83.620 ku	10.850	90,600 ku	10,720	12.0	782	3,600	2000(est)	952	
dins -ma	1980	99.870 ku	10.470	107,900 ku	10,400	14.0	611	3,600	2060(est)	1029	
Mu-7018	1976	90,830 ku	11,160	97,780 84	11,070	11.0	863	3,000	2064(est)	1166	
Mitsul Engl	ineering 6 Shipbui	lding									
SB-90C	1969	17,500 84	13,020	19,300	12,750	4.6	175	5,475	1940	1060	
Nuovo Pign	ne										
			073 61			7.2	215	6,500	1730	980	General Electric License
MS 3142	6461	10,200 10	000 01	24 600 14	12 180	10.2	24.5	5,100	1805	056	
MS 2001	9261	NA 057 CL	11.150	34.700 ku	11.190	11.5	304	5,100	1950	952	
1000	1970	75.000 11	10.590	80.240 ku	10,670	11.5	104	3,600	2085	1035	
1006 SM	1976	105,600 kM	10,700	113,300 ku	10,770	9.6	876	3,000	1950	1010	
Pratt 6 Wh	itney Aircraft Can	ada									
							101	1 600	1970	877	
CFT4C-3F	1976	29,820 kM	10,880	32, 180 kW	10,760	13.8	900	3.600	1970	877	
PLAC	1978	NN 056'65	10,820		10,100	0.11					

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TABLE 1.2.2.1-1 UPERATING CHARACTERISTICS OF CUMMERCIALLY AVAILABLE GAS TURBINES ("ant'd) Foreign Manufacturers

	Year	and a	Lond	Peak L	pad				Tenperat	Tre'F	Comments
	First		Hear Sure		Heat Rate	Pressure	Flow Rate	Speed	Turbine		
Model	Appear ance	2nd Lees	Btu/killi	Output	Buu/Rikh	Ratio	ib/sec	al.	Inlet	Exhaust	
Rolls-Royce Indu	strial and Marine										
Sper	1976	NY 006'11	10,240	13,800 ku	10,050	0.91	128	1	1	1	
Avon 1513	1964	12,000 ku	12,490	15,600 ku	11,650	10.0	180	1	1	1	
Avon 1514	1970	15,600 ku	11,770	17,400 84	11,480	10.0	180		1	1	
Avon 1515	\$161	16, 300 ku	11,790	18,200 kV	11,540	0.01	181			1	
KB 211	1974	23,800 84	6,840	-	-	19.2	201				
01 ympus	1962	26,100 44	11,040	28,000 44	10.980	0.11	240	3000/ 3600	1990 est.	1024	
SK.30	1161	26,100 ku	070'11	28,000 84	096.01	0.11		0006/0006	1	200	
SM60	1411	22,200 44	040 11	NA 000,00	10,980	0.11		0000 M000		006	
1001	*641			· · · · · · · · · · · · · · · · · · ·	000-1-11						
Stal-Laval Turbi	c										
	194.0	13 550 54	11 610	12 250 40	11 500	0 21	186	10001 1000	1560	140	
PP1-400m 1511	1044	14 100 11	12 220	115, 400, 11	12.200	10.0	180	3000/3600	1560	999	
GT120C	1939	70,000 14	11.960	75,000 kv	11.780	15.0	061	3000	1530	635	
GT 200	1977	75,000 ku	10,300	NN 000 58	10,120	16.0	770	3000	1	1	
Sulter Brothers											
Type J	1976	5, 300 ku	12,640	5,550 kV	12,545	8.1	62	10,000	1750	916	
Type R3	1976	5.150 kM	10,340	5,400 ku	10,155	8.3	62	10,000	1750	920	
Type 7	1970	10,580 ku	13,485	11,100 84	13,380	5.7	101	9,400	1700	969	
Type MJ	1970	10.020 kM	11,190	10,530 kW	10.610	9.7	111	007.9	00/1	006	
TELED 12	1978	11 700 40	071 61			0.01	167	005.9	1775 • • •	925	
PRIMO 15	1978	14. 900 14	12.015			10.0	169	6.500	1850 est.	870	
PRIMO 21	1979	21,000 ku	10.215		-	20.0	1946	6.500	1	887	
	1										
Thomassen Hollan	21										
C31423	6761	10.450 km	13,320			7.2	115	6.500	1730	619	
G3142R	1561	10,000 ku	10,370	-		1.2	115	6.500	1730	899	
G5 141P	1958	25, 350 84	12,020			10.2	269	5,100	1730	106	
PGS J4.1P	1961	24,500 ku	12,310	26,450 kW	12,220	10.2	266	5,100	1805	556	
PC6441A	1979	11, 750 ku	11,260	34.750 kH	11,210	1.11	100	5,100	1850	106	County Planets land
PC64418	1961	14, 140 ku	11,280	39°,900 KM	067,11		105	001.0	1970	104	
W/1016	1919	10,400 to	010101	00 000 00	10 620		100	0071	5861	477	
COLLE	1979	N 00, 101	10.420			9.0	876	3.000	1955	655	
3141604	6/61	10, 500 14	10,820	113, 370 ku	10,790	9.6	898	3,000	2055	1020	
Toshiba											
Twee 9	1970	N. 500 kW	12,010	37,500 84	11,890	6.9	348	4473/4485	1	515	
Type 11	1/61	72,500 10	19,760	78,900 44	10,660	11.0	626	3660		975	Rrown Boveri License
Type 13	0/61	79.400 ku	11,370	86,900 ku	11,220	10.4	661	3000		\$05	
west inghouse											
	1 44.0	18 010 10	11.220	14 150 64	11 010	6.1	273	4912	1450	273	
01416C	1 966	17 200 44	13.390	19,000 14	13.210	1.0	270	2165	1450	173	
w191PG Super	1980	19. 250 kW	12.910			8.6	297	4912	1450	758	
w3010	1961	32, 300 14	13,060	34,400 84	12,870	6.8	438	3600	1450	800	
w301G Super	8/61	35, 125 24	12,600	1		1.5	187	3400	1450	785	

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TABLE 1.2.2.1-2 OPERATING CHARACTERISTICS OF COMPERCIALLY AVAILABLE COMPLIED CYCLE PLANTS

	Year	Base	bead	Peak	Load			Test inc		Engine No.
odel	First Appearance	Output	Btu/kith	Output	Btu/kuh	Frequency	Type	Size	Steas	and Type
and the second se									10 6001 601	1311
OCY 50	1979	21,000 14	8.520	24,400 14	0/7.9	24 15	non-fired	10 000 FM	27.500 14	two 9251
OCI 100	141	10 Not 10	0.000		011 1		non-fired	91.300 84	37.000 ku	one ¥1101
001 100	6261	14.3,000 64	044.1	153,000 14	7,430	20 Hz	non-tired	91, 700 ku	51, 300 ku	one W1101
Isthon-Atlantique										
FCA 105P	1976	13,100 ku	8,850	36,000 ku	8,570	50/60 Hz	non-fired	22,400 ku	10,700 ku	one PG5341
EGA 205P	1976	69,000 kM	8, 575		I	50/60 Hz	non-fired	47.350 km	22,100 64	176534 OA1
EGA 206P	8/61	102, 300 kM	9,030	1	I	2H 09/00	non-fired			
CCA 2000	1976	2.17.000 14	7.900	1	1	50 Hz	non-fired	168,900 ku	10,000 64	tun PC9111
100 YOOL	1978	290,000 M	161 1		1	50 Hz	non-fired	209,400 ku	81,500 ku	two Pr.9141
EGA 109E	1978	144,200 km	2.845		I	2H 05	non-fired	104.700 84	10,000 10	one PC9141
BC Brown, Bover	b Co. Ltd									
rown. Bovert &	(Le AG									
LOWN BOVELL INT	Tranto Page									
1-64	1977	51,100 14	8,105	55,690 ku	8.025	50/60 Hz	non-fired	31.870 ku	19.230 km	one Type 9
X9-2	1977	101,990 ku	7,955	113, 350 ku	1.875	50/60 Hz	non-fired	07. 140 M	N 000 11	two Type 9
19-3	6261	157,210 ku	516.1	171, 350 64	(18.1	ZH 09/00	Dell'I-uou	127 480 ku	82.500 10	four Tune 9
4-6Y	1979	209,980 ku	1,880	778.810 40	< 09.1	78 00/00	Patt: -000			a addit more
	19.79	106. 790 ku	7.435	114.200 14	7.360	50 Hz	non-fired	68,400 ku	36, 390 ku	one Type II
1-11-2	1979	211.550 ku	7.355	230,600 ku	7,285	50 Hz	non-fired	136,800 ku	14.750 84	three Type 11
11-3	1979	317, 320 64	7,350	345,880 kV	1,280	2H 09	non-tired	205,200 ku	112,120 14	four Type 11
4-11v	1979	426, 380 ku	7, 305	464,700 kM	7.235	50 Hz	pon-tired			
	1979	111 660 14	7.200	123.880 kV	7.630	SO Hz	non-fired	77,160 ku	36,500 84	one Type 13
1-11-2	1979	228, 150 ku	7.670	248,900 84	1,600	SO Hz	non-fired	154, 320 ku	74, 030 ku	these Type 13
1-11-1	6161	344, 050 ku	7,635	375,000 44	7.565	50 Hz	non-fired	231.480 kW	112.570 10	four Type 13
2-CIN	6/61	459,800 kW	7.615	501,200 kW	7.545	20 Hz	non-fired	1049'800		
urtiss-Wright P	over Systems									
		and then you		100 000 000	0,000	Sulad No.	non-fired	89.000 ku	61,000 kV	four
051 331	1978	000 001	001.6	100,000 m c	000.6	20/60 Hz	non-11 red	NN 000.68	121,000 ku	four
TEC 200	1978	210,000 84	nof *4	an opping						
Figt. TTG										
		111 100 111				50/60 Hz	non-tired	36, 300 MM	17,400 44	one TG.20
00 20	1978	101 400 401	111.8	1		50/60 Hz	non-fired	72,600 ku	34 800 KM	two TG. 20
CC 100	1978	126,000 14	8.015	1		50 Hz	non-fired	85,000 ku	38,000 ku	one TG.50
CC 530	1978	252,000 44	8,035		•	2H 05	non-tired	176,000 ku	76,000 84	two TC. 50
and the state of the										
ALL DAY INTOTHE										
EM810	1975	118,200 44	1, 790	1		20 Hz	part 1-uou	1		
General Electric	CT DIV									
		00 1001	070 6	1		50 Hz	non-fired	1	1	one MS9001E
STAG 109E	111	140, 400 LU	1 890		-	50 Hz	non-fired		-	Two MS9001E
STAG 209E	10.01	271 and 10	7.890		-	20 Hz	non-fired		1	three MS9001E
STAC DUSE	1479	564. 700 ku	7,890	1		50 Hz	non-fired		1	tour MS9001E
					Ì	1		T.	E	1
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TABLE 1.2.2.1-2 OPERATING CHARACTERISTICS OF COMMENCIALLY AVAILABLE COMBINED CYCLE PLANTS (Cont'd)

	Year	A150	Load	Park In	1					
Model	First Appearance	Output	Heat Rate Stu/kUh	Output	Heat Rate Btu/bHh	Output	Botter	Turbine	Stean	Engine No. and Type
Stal-Laval_Tu	rbin									
Gastean 125 Gastean 250	1977	113.000 ku	7.420	122,000 ku	7.420	50 Hz	non-fired non-fired	80,000 kV	34,000 ku	one 61.00 two 61200
Sul zer Grothe	5									
Turbotur 305	1974	15, 300 80	8,210	-		50/60 Hz	non-fired	10.600 ku	5.700 80	two Type 3
Turbotur 210	7/61	17.000 14	8, 340	-	ļ	50/60 Hz	non-fired	10.580 84	6.420 kg	One Type 7
Turbotur 310	1974	34,000 kV	8, 110	-		20/60 Hz	non-fired	21,160 44	12,840 ku	Two Type 7
Turbotur 214	1978	20.400 14	8,030	-		50/40 Hz	non-fired	13,700 14	6,700 ku	One PRIMO 14
Turbotur 314	8/61	11,000 14	8,010		-	20/60 Hz	non-fired	27,400 84	13,600 84	Two PRING 14
Turbotur 325	8/61	78,800 84	7,300			50/60 Hz	non-fired	25,400 ku	13,900 ku 28,000 ku	One Olympus Two Olympus
Thomassen Hol	pur									
STEG 105P	1972	11,400 ku	8.740			50/60 Hz	non-flrod			One PC5141
STEG 205P	1972	67.200 ku	8,720	1		50/60 Hz	non-fired	1	1	Two PGSJ41
STEG 1068	1980	47,800 44	8,120			50 Hz	non-fired	1	1	One PC6441
STEC 2068	1980	95, 900 ku	8,100			50 Hz	non-fired	1	-	Two PG6441
STEC 107E	6191	99, 300 ku	7,700	-	1	60 Hz	non-f)red	1	1	One PC7891
STEC 207E	6261	198,800 ku	7,660			50 Hz	nun-fired	1	-	Two PG7891
STEG 109E	1979	140.400 84	7.940			50 Hz	mon-fired		-	One PG9141
STEG 209E	6261	282, 300 kM	7,840	-		50 Hz	non-fired	-	1	Two P69141
UTC Power Sys	i un									
2-451	1980	81,150 84	8. 385			40 Hz	non-fired	NA 050.75	21.900 64	Two FTLC
4-451	0861	162,950 84	8, 350			50 Hz	non-fired	114. 500 ku	48.450 kv	Four F14C
15P-8	1411	326,580 ku	8, 135	-		60 Hz	non-fired	229,000 ku	97,580 ku	eight FT4C
Hest Inghouse	anada									
Mini Pace	1981	51,445 kW	9.800	I		50/60 Hz	bunker	14.244 NU	18,200 ku	Two w191 both gas
Vest Inchouse	pabustion Turb	ine Systems Div.								
PACE 2511	1961	55, 200 44	8,300	****	****	\$0/60 Hz	non-fired	NA 008.AU	18.400 ku	one W2518
PACE 2512	1861	110.400 bu	8, 300		-	50/60 Hz	non-fired	73,500 ku	36.900 ku	two W2518
PACE 2514	1861	220,800 84	8, 300			ZH 09/05	non-fired	110, 300 ku	55, 300 ku	three V2518
			-			71 04/01	B3111-000		-	
PACE 5011	1861	134,700 ku	7.430			\$0/60 Hz	non-fired	93,800 ku	40.900 ku	one ¥5010
PACE 5012	1981	269,400 49	7.630	-	-	50/60 Hz	non-fired	187,500 MM	81, 900 ku	Two #5010
PACE 2013	1961	404,100 kW	7.630			20/60 Hz	non-fired	281, 300 ku	122.800 44	three wolld
PACE 3015	1861	671,600 ku	7.430			50/60 Hz 50/60 Hz	non-fired	375,000 44	163.800 MV	four WS010
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TABLE 1.2.2.1-2 OPERATING CHARACTERISTICS OF CONDERCIALLY AVAILABLE CONDINED CYCLE PLANTS (Cont'd)

Output Iteluency Type Size Stem And Type	Oxfert Ixu/Mi. Frequency Jpc Size Scens and Jpc	Output Jack Mod. Terpuned Jack State State And Jack	Obtoolid MarkNall Expresso April	
30/60 Hz con-fired con-fired for W \$5001 <t< th=""><th> 0//0 NL constitued for NS001 9//0 NL constitued constitued constitued constitued constitued constitued constitued constitued constitued constitued constitued constitued</th><th> (ever 16500) </th><th></th><th>ance Output Btu/kWh</th></t<>	0//0 NL constitued for NS001 9//0 NL constitued constitued constitued constitued constitued constitued constitued constitued constitued constitued constitued constitued	(ever 16500)		ance Output Btu/kWh
	S9700 Hz Non-fired Non Ker (100) fore K5001 fore K5001 fore K5001 fore K5001	Image: constrained by the constraned by the constrained by the constrained by the con		000 B 779 007 B11 1
				0. 200 km 8, 070
Image: constraint of the constrant of the constraint of the constraint of the constraint of the c	00 Nz con-1/red 0 Nz con-1/red 0 nz 15,001 0 nz 15,001 15,001 15,001 15,001 <t< td=""><td> 00 Rt mon-fired ore R51001 71,400 LW 6,100 LW 6,100 LW 72,100 LW 72,100 LW 66,100 LW 72,100 LW 66,100 LW 72,100 LW 66,100 LW <td< td=""><td> 00 lt constitued onstitued constitued <thconstitued< th=""> <thconstitued< th=""></thconstitued<></thconstitued<></td><td>9 192,400 km 8,070</td></td<></td></t<>	00 Rt mon-fired ore R51001 71,400 LW 6,100 LW 6,100 LW 72,100 LW 72,100 LW 66,100 LW 72,100 LW 66,100 LW 72,100 LW 66,100 LW <td< td=""><td> 00 lt constitued onstitued constitued <thconstitued< th=""> <thconstitued< th=""></thconstitued<></thconstitued<></td><td>9 192,400 km 8,070</td></td<>	00 lt constitued onstitued constitued <thconstitued< th=""> <thconstitued< th=""></thconstitued<></thconstitued<>	9 192,400 km 8,070
	1 00 Hz mon-fired mon-fired	The second sec	0.01 concrited 10.000 M 27.000 M	7 00 km 7 200
13, 400 14 00m-fired 0m-fired 0	15,400 kt 6,100 kt ann-fired 46,000 kt 21,000 kt 51,000 kt 50,000 kt <th< td=""><td></td><td>75.400 M 5.3 (0) 0 (R concrited 0.0 (R <thcoccrited< th=""> 0.0 (R concrited</thcoccrited<></td></th<> <td>394, 700 km 7, 650</td>		75.400 M 5.3 (0) 0 (R concrited 0.0 (R <thcoccrited< th=""> 0.0 (R concrited</thcoccrited<>	394, 700 km 7, 650
34,400 kt 8,300 50/60 kt mon-fired 56,000 kt 27,000 kt 57,000 kt 56 681 27,000 kt 8,120 50 kt mon-fired 35,000 kt 57,000 kt 56 682 27,000 kt 8,120 50 kt mon-fired 117,000 kt 27,000 kt 57,000 kt 56 662 291,000 kt 8,110 50 kt mon-fired 117,000 kt 116,000 kt 57,000 kt 56 6782 291,000 kt 7,340 60 kt mon-fired 117,000 kt 116,000 kt 57,000 kt 56 6782 2111,000 kt 7,340 60 kt mon-fired 117,000 kt 110,000 kt 57,000 kt 57,001 kt	71,400 kt 5,900 50/60 kt mon-fired 46,000 kt 22,000 kt 50/60 kt <t< td=""><td>3,400 kt 6,100 kt 5,100 kt 00/60 kt mon-fired 3,000 kt 27,000 kt mon-fired 3,000 kt 1,000 kt</td><td>3.4.00 kt 5.900 kt mon-fired on transfered on transfered</td><td>9 595,600 kM</td></t<>	3,400 kt 6,100 kt 5,100 kt 00/60 kt mon-fired 3,000 kt 27,000 kt mon-fired 3,000 kt 1,000 kt	3.4.00 kt 5.900 kt mon-fired on transfered on transfered	9 595,600 kM
74,400 kt 8,300 kt 50 kt mon-fired 46,000 kt 27,000 kt tee FG381 91,000 kt 8,110 60 kt mon-fired 117,000 kt 57,000 kt tee FG381 91,000 kt 8,110 60 kt mon-fired 17,000 kt 57,000 kt tee FG381 91,000 kt 8,110 60 kt mon-fired 17,000 kt 57,000 kt tee FG381 111,000 kt 7,900 kt 7,900 kt 17,000 kt 17,000 kt tee FG381 111,000 kt 7,900 kt 7,900 kt 17,000 kt tee FG381 111,000 kt 7,900 kt 17,000 kt tee FG381 111,000 kt 7,900 kt 17,000 kt tee FG381 111,000 kt 7,900 kt 117,000 kt tee FG381 112,000 kt 8,110 90 kt mon-fired 21,000 kt tee FG381 113,000 kt 8,110 90 kt mon-fired 21,000 kt tee FG381 114,000 kt 8,110 90 kt mon-fired 21,000 kt tee FG381 118,000 kt 8,100 kt 8,100 kt 114,000 kt tee FG381 <td>74,400 kt 8,330 50/60 kt mon-fired 35,000 kt 22,300 kt mon Fired 35,000 kt 57,000 kt mon Fired 35,000 kt mon Fired 37,000 kt mon Fired 32,000 kt mon Fired 32,000 kt mon Fired 33,000 kt mon Fired 33,000 kt mon Fired 32,000 kt mon Fired 33,000 kt mon Fired 33,000 kt mon Fired 32,000 kt mon Fired 32,000 kt mon Fired 32,000 kt</td> <td>74,400 kt 8,330 90/60 kt con-fired 39,000 kt 22,300 kt con-6731 73,000 kt 8,130 60 kt con-fired 117,000 kt 77,000 kt two-6731 73,1000 kt 8,130 60 kt con-fired 117,000 kt 77,000 kt two-6731 73,1000 kt 7,800 kt 17,000 kt 17,000 kt 137,000 kt 10,000 kt two-6731 111,000 kt 7,800 kt 00 kt con-fired 137,000 kt 137,000 kt two-6731 131,000 kt 7,800 kt 17,000 kt 137,000 kt 137,000 kt two-67381 131,000 kt 8,110 90 kt con-fired 29,000 kt 137,000 kt two-67381 131,000 kt 8,110 90 kt con-fired 29,000 kt two-67381 132,000 kt 8,110 90 kt con-fired 29,000 kt two-67381 131,000 kt 8,110 90 kt con-fired 29,000 kt two-67381 132,000 kt 8,110 90 kt con-fired 29,000 kt 114,00 kt con-71381 148,000 kt 8,110</td> <td>3.400 kg 5.700 kg</td> <td></td>	74,400 kt 8,330 50/60 kt mon-fired 35,000 kt 22,300 kt mon Fired 35,000 kt 57,000 kt mon Fired 35,000 kt mon Fired 37,000 kt mon Fired 32,000 kt mon Fired 32,000 kt mon Fired 33,000 kt mon Fired 33,000 kt mon Fired 32,000 kt mon Fired 33,000 kt mon Fired 33,000 kt mon Fired 32,000 kt mon Fired 32,000 kt mon Fired 32,000 kt	74,400 kt 8,330 90/60 kt con-fired 39,000 kt 22,300 kt con-6731 73,000 kt 8,130 60 kt con-fired 117,000 kt 77,000 kt two-6731 73,1000 kt 8,130 60 kt con-fired 117,000 kt 77,000 kt two-6731 73,1000 kt 7,800 kt 17,000 kt 17,000 kt 137,000 kt 10,000 kt two-6731 111,000 kt 7,800 kt 00 kt con-fired 137,000 kt 137,000 kt two-6731 131,000 kt 7,800 kt 17,000 kt 137,000 kt 137,000 kt two-67381 131,000 kt 8,110 90 kt con-fired 29,000 kt 137,000 kt two-67381 131,000 kt 8,110 90 kt con-fired 29,000 kt two-67381 132,000 kt 8,110 90 kt con-fired 29,000 kt two-67381 131,000 kt 8,110 90 kt con-fired 29,000 kt two-67381 132,000 kt 8,110 90 kt con-fired 29,000 kt 114,00 kt con-71381 148,000 kt 8,110	3.400 kg 5.700 kg	
97,000 ku 6,200 ku 57,000	91,000 M. S.100 M. S.000 M. M. M. M. M. <	91,000 M Ammetted 31,000 M Constrained 117,000 M<	97,000 M 6,130 00 M mon-fired 11,000 M 27,000 M 57,000 M for CRR1 97,000 M 6,130 00 M mon-fired 11,000 M 27,000 M for CRR1 97,000 M 7,100 M 00 M mon-fired 13,000 M for CRR1 111,000 M 7,100 M 00 M mon-fired 13,000 M for CRR1 111,000 M 8,110 00 M mon-fired 13,000 M for CRR1 111,000 M 8,110 00 M mon-fired 13,000 M for CRR1 111,000 M 8,110 00 M mon-fired 13,000 M for CRR1 111,000 M 8,110 00 M mon-fired 13,000 M for CRR1 118,000 M 8,110 00 M mon-fired 15,000 M for CRR1 118,000 M 8,110 00 M mon-fired 15,000 M for CRR1 118,000 M 8,110 115,000 M 115,000 M for CRR1 118,000 M 8,100 M 115,000 M 10,000 M	0 58, 300 NU 9,110
17,000 kv 8,120 60 kk mon-fired 11,000 kv 8,000 kv few PC381 271,000 kv 8,110 60 kk mon-fired 235,000 kv 13,000 kv few PC381 115,000 kv 7,300 kv 1,300 kv 13,000 kv 50 kc mon-fired 13,000 kv 50 kc 115,000 kv 7,300 kv 7,300 kv 13,000 kv 51,000 kv 50 kc 50 kc 131,000 kv 7,300 kv 17,000 kv 117,000 kv 50 kc 50 kc 131,000 kv 8,210 50 kc mon-fired 220,000 kv 101,000 kv 50 kc 131,000 kv 8,110 50 kc mon-fired 231,000 kv 50 kc 50 kc 131,000 kv 8,110 50 kc mon-fired 13,000 kv 50 kc 50 kc 131,000 kv 8,110 50 kc mon-fired 232,000 kv 50 kc 50 kc 131,000 kv 8,110 50 kc 15,000 kv 137,000 kv 50 kc 50 kc 131,000 kv 8,110 50 kc 15,000 kv 15,000 kv 50 kc 50 kc 131,000 kv 8,110 50 kc 15,000 kv 15,000 kv 50 kc 50 kc 148,000 kv 8,110 50 kc	17,000 kt 6,100 kt 60 kt mon-fired 11,000 kt 5,000 kt free 7/31 91,000 kt 8,110 60 kt mon-fired 13,000 kt 5,000 kt free 7/31 91,000 kt 7,940 60 kt mon-fired 13,000 kt 6 kt 7/31 111,000 kt 7,940 60 kt mon-fired 13,000 kt 6 kt 7/31 131,000 kt 1,200 kt 13,000 kt 13,000 kt 10,000 kt 10,000 kt 10,000 kt 132,000 kt 8,110 30 kt mon-fired 23,000 kt 11,000 kt 10,000	713,000 kv 8,110 60 kv mon-fired 115,000 kv 93,000 kv ber FC3R1 713,000 kv 7,900 kv 60 kv mon-fired 13,000 kv 93,000 kv ber FC3R1 713,000 kv 7,800 kv 60 kv mon-fired 13,000 kv b1,000 kv b7,000 kv <td>17,000 M 6,130 0.01 meetired 11,000 M 5,000 M fear F331 11,000 M 5,100 M 0.01 meetired 13,000 M 0.00 M meetired 13,000 M meetired 14,000 M meetired 14,000 M 14,000 M</td> <td>0 86,000 14 8,430</td>	17,000 M 6,130 0.01 meetired 11,000 M 5,000 M fear F331 11,000 M 5,100 M 0.01 meetired 13,000 M 0.00 M meetired 13,000 M meetired 14,000 M meetired 14,000 M 14,000 M	0 86,000 14 8,430
97,000 kt 8,110 60 kt mm-fired 233,000 kt 10,000 kt fear PG381 211,000 kt 7,800 kt 60 kt mm-fired 11,000 kt 57,000 kt fear PG381 211,000 kt 7,800 kt 60 kt mm-fired 147,000 kt 57,000 kt fear PG381 211,000 kt 7,800 kt 7,800 kt 60 kt mm-fired 29,000 kt 10,000 kt fear PG381 213,000 kt 8,210 30 kt mm-fired 29,000 kt 11,000 kt fear PG381 213,000 kt 8,210 30 kt mm-fired 29,000 kt 10,000 kt fear PG381 218,000 kt 8,110 30 kt mm-fired 15,000 kt 10,000 kt fear PG381 218,000 kt 8,110 30 kt mm-fired 15,000 kt 10,000 kt fear PG381 218,000 kt 8,110 30 kt mm-fired 15,000 kt 10,000 kt fear PG381 218,000 kt 8,110 30 kt mm-fired 15,000 kt 15,000 kt fear PG381 218,000 kt 8,110 30 kt mm-fired 15,000 kt 15,000 kt fear PG381 218,000 kt 8,100 kt 10,000 kt 115,000 kt 11,000 kt fear PG311 <td>73,000 K Mon-fired 235,000 K For F7381 111,000 1,000 1,000 0 M Mon-fired 13,000 M Mer F7381 111,000 1,000 1,000 M M Mon-fired 13,000 M Mer F7381 111,000 1,000 M Mon-fired 10,000 M Mon Fired 13,000 M Mer F7381 111,000 M 8,210 90 M Mon-fired 293,000 M Mer F7381 1118,000 M 8,110 90 M Mon-fired 293,000 M Mer F7381 118,000 M 8,110 90 M Mon-fired 293,000 M Mer F7381 118,000 M 8,110 90 M Mon-fired 193,000 M Mer F7381 118,000 M 8,100 M 135,000 M Mer F7381 118,000 M 8,100 M 13,00</td> <td>97,000 M 6,110 60 M mon-fired 23,000 M 15,000 M four FC/961 231,000 M 7,800 60 M mon-fired 13,000 M 60 M four FC/961 231,000 M 7,800 60 M mon-fired 23,000 M 13,000 M four FC/961 231,000 M 8,110 50 M mon-fired 23,000 M 13,000 M four FC/961 237,000 M 8,110 50 M mon-fired 23,000 M 13,000 M four FC/961 278,000 M 8,110 50 M mon-fired 13,000 M four FC/961 278,000 M 8,110 50 M mon-fired 13,000 M four FC/961 278,000 M 8,110 50 M mon-fired 13,000 M four FC/961 278,000 M 8,110 50 M mon-fired 13,000 M four FC/961 278,000 M 8,110 50 M 11,15,000 M four France four France 279,000 M 60 M 11,15,000 M 11,15,000 M four France four France 279,000 M 11,15,000 M 11,15,000 M 11,15,000 M four France 270,000 M 10,110 M 60 M 11,15,000 M four France 270,000 M 10,110 M 11</td> <td>0.1.000 0.01 constituted 13,000 10 const</td> <td>001 8 000 00 00 000 000 000 000 000 000</td>	73,000 K Mon-fired 235,000 K For F7381 111,000 1,000 1,000 0 M Mon-fired 13,000 M Mer F7381 111,000 1,000 1,000 M M Mon-fired 13,000 M Mer F7381 111,000 1,000 M Mon-fired 10,000 M Mon Fired 13,000 M Mer F7381 111,000 M 8,210 90 M Mon-fired 293,000 M Mer F7381 1118,000 M 8,110 90 M Mon-fired 293,000 M Mer F7381 118,000 M 8,110 90 M Mon-fired 293,000 M Mer F7381 118,000 M 8,110 90 M Mon-fired 193,000 M Mer F7381 118,000 M 8,100 M 135,000 M Mer F7381 118,000 M 8,100 M 13,00	97,000 M 6,110 60 M mon-fired 23,000 M 15,000 M four FC/961 231,000 M 7,800 60 M mon-fired 13,000 M 60 M four FC/961 231,000 M 7,800 60 M mon-fired 23,000 M 13,000 M four FC/961 231,000 M 8,110 50 M mon-fired 23,000 M 13,000 M four FC/961 237,000 M 8,110 50 M mon-fired 23,000 M 13,000 M four FC/961 278,000 M 8,110 50 M mon-fired 13,000 M four FC/961 278,000 M 8,110 50 M mon-fired 13,000 M four FC/961 278,000 M 8,110 50 M mon-fired 13,000 M four FC/961 278,000 M 8,110 50 M mon-fired 13,000 M four FC/961 278,000 M 8,110 50 M 11,15,000 M four France four France 279,000 M 60 M 11,15,000 M 11,15,000 M four France four France 279,000 M 11,15,000 M 11,15,000 M 11,15,000 M four France 270,000 M 10,110 M 60 M 11,15,000 M four France 270,000 M 10,110 M 11	0.1.000 0.01 constituted 13,000 10 const	001 8 000 00 00 000 000 000 000 000 000
11,000 ku 7,940 60 kt con-fired 71,000 ku 71,000 ku con-fired 71,000 ku con-fired <t< td=""><td>11.1,000 kt 7,940 60 kt mon-fired 73,000 kt 57,000 kt</td><td>11,000 ku 7,940 60 ft mon-fired 13,000 ku 57,000 ku 57,900 ku 57,961 31,000 ku 7,800 60 ft mon-fired 233,000 ku 57,000 ku 57,961 31,000 ku 7,800 60 ft mon-fired 233,000 ku 57,000 ku 57,961 31,000 ku 8,110 50 ft mon-fired 233,000 ku 57,000 ku 50,000 ku 57,961 278,000 ku 8,110 50 ft mon-fired 13,000 ku 4,500 ku 50,000 ku<td>11,000 km 7,900 km 60 km con-fired 13,000 km 13,000 km ree PC391 211,000 km 7,800 km 60 km con-fired 10,000 km 10,000 km ree PC391 211,000 km 7,800 km 60 km con-fired 29,000 km 10,000 km ree PC391 218,000 km 8,110 90 km con-fired 15,000 km 10,000 km ree PC391 218,000 km 8,110 90 km con-fired 15,000 km 115,000 km ree PC311 218,000 km 8,110 90 km con-fired 15,000 km 15,000 km ree PC311 218,000 km 8,110 90 km con-fired 15,000 km 10,000 km ree PC311 218,000 km 8,110 90 km con-fired 21,500 km 10,000 km ree PC311 218,000 km 8,110 90 km con-fired 21,500 km 10,400 km ree PC311 218,000 km 91,600 km 91,600 km 10,100 km 10,400 km ree PC311 218,000 km</td><td>067 8 AV 000 MM 0 200</td></td></t<>	11.1,000 kt 7,940 60 kt mon-fired 73,000 kt 57,000 kt	11,000 ku 7,940 60 ft mon-fired 13,000 ku 57,000 ku 57,900 ku 57,961 31,000 ku 7,800 60 ft mon-fired 233,000 ku 57,000 ku 57,961 31,000 ku 7,800 60 ft mon-fired 233,000 ku 57,000 ku 57,961 31,000 ku 8,110 50 ft mon-fired 233,000 ku 57,000 ku 50,000 ku 57,961 278,000 ku 8,110 50 ft mon-fired 13,000 ku 4,500 ku 50,000 ku <td>11,000 km 7,900 km 60 km con-fired 13,000 km 13,000 km ree PC391 211,000 km 7,800 km 60 km con-fired 10,000 km 10,000 km ree PC391 211,000 km 7,800 km 60 km con-fired 29,000 km 10,000 km ree PC391 218,000 km 8,110 90 km con-fired 15,000 km 10,000 km ree PC391 218,000 km 8,110 90 km con-fired 15,000 km 115,000 km ree PC311 218,000 km 8,110 90 km con-fired 15,000 km 15,000 km ree PC311 218,000 km 8,110 90 km con-fired 15,000 km 10,000 km ree PC311 218,000 km 8,110 90 km con-fired 21,500 km 10,000 km ree PC311 218,000 km 8,110 90 km con-fired 21,500 km 10,400 km ree PC311 218,000 km 91,600 km 91,600 km 10,100 km 10,400 km ree PC311 218,000 km</td> <td>067 8 AV 000 MM 0 200</td>	11,000 km 7,900 km 60 km con-fired 13,000 km 13,000 km ree PC391 211,000 km 7,800 km 60 km con-fired 10,000 km 10,000 km ree PC391 211,000 km 7,800 km 60 km con-fired 29,000 km 10,000 km ree PC391 218,000 km 8,110 90 km con-fired 15,000 km 10,000 km ree PC391 218,000 km 8,110 90 km con-fired 15,000 km 115,000 km ree PC311 218,000 km 8,110 90 km con-fired 15,000 km 15,000 km ree PC311 218,000 km 8,110 90 km con-fired 15,000 km 10,000 km ree PC311 218,000 km 8,110 90 km con-fired 21,500 km 10,000 km ree PC311 218,000 km 8,110 90 km con-fired 21,500 km 10,400 km ree PC311 218,000 km 91,600 km 91,600 km 10,100 km 10,400 km ree PC311 218,000 km	067 8 AV 000 MM 0 200
231,000 ki 7,630 60 ki non-fired 147,000 ki 67,000 ki tee PG781 332,000 ki 7,800 60 ki non-fired 220,000 ki 137,000 ki tee PG781 439,000 ki 8,210 50 ki non-fired 15,000 ki 40,000 ki tee PG911 418,000 ki 8,110 50 ki non-fired 15,000 ki 125,000 ki tee PG911 418,000 ki 8,110 50 ki non-fired 21,500 ki 125,000 ki tee PG911 418,000 ki 8,110 50 ki non-fired 21,500 ki 125,000 ki tee PG911 418,000 ki 8,110 50 ki non-fired 21,500 ki 125,000 ki tee PG911 418,000 ki 8,110 50 ki non-fired 21,500 ki 125,000 ki tee F7386 50,600 ki non-fired 21,500 ki 125,000 ki tee F7386 50,600 ki non-fired 21,500 ki 12,500 ki tee F7386 50,600 ki non-fired 21,500 ki 11,400 ki 11,400 ki tee F7386 50,600 ki non-fired 21,500 ki 14,500 ki tee F7386 50,600 ki non-fired 21,500 ki tee F7386 51,600 ki 24,100 ki tee F7386 51,600 ki 241,200 ki tee F7386 50,600 ki 1500 ki 16,400 ki tee F7386 50,600 ki 16,000 ki 16,400 ki tee F7386 50,600 ki 16,000 ki tee F7386 50,600 ki tee F7386 50,000 ki 16,000 ki 16,000 ki 16,000 ki tee F7386 50,000 ki 16,000 ki tee F7386 50,000 ki 16,000 ki 16,000 ki 16,000 ki tee F7386 50,000 ki 16,000 ki 16,000 ki 16,000 ki tee F7386 50,000 ki 16,000 ki 16,0	211,000 Lk 7,800 Lk 60 Hk non-fired 147,000 Lk 67,000 Lk forer PG361 352,000 Lk 7,800 Lk 60 Hk non-fired 230,000 Lk 137,000 Lk forer PG361 128,000 Lk 8,110 30 Hk non-fired 15,000 Lk 40,000 Lk forer PG3611 278,000 Lk 8,110 30 Hk non-fired 15,000 Lk 40,000 Lk forer PG3611 278,000 Lk 8,110 30 Hk non-fired 15,000 Lk 155,000 Lk forer PG3011 278,000 Lk 8,110 30 Hk non-fired 155,000 Lk forer PG3011 278,000 Lk 8,110 30 Hk non-fired 15,500 Lk forer F73611 278,000 Lk 8,110 30 Hk non-fired 11,500 Lk forer F7366 111 10,000 Lk 4,500 Lk 11,4,50 Lk forer F7366 111 11,4,50 Lk 11,4,50 Lk forer F7366 forer F7366 111 11,4,50 Lk 11,4,50 Lk forer F7366 forer F7366 111 11,4,50 Lk 11,4,50 Lk forer F7366 forer F7366 111 11,4,50 Lk 11,4,50 Lk forer F7366 forer F7366 111 11,1,1,10 Lk 11,4,50 Lk f	211,000 kt 7,800 kt 60 kt mon-fired 147,000 kt 57,000 kt four FC3981 053,000 kt 7,800 kt 60 kt mon-fired 220,000 kt 113,000 kt four FC3981 128,000 kt 8,210 30 kt mon-fired 15,000 kt 100,000 kt four FC3981 228,000 kt 8,110 30 kt mon-fired 15,000 kt 10,000 kt four FC3981 278,000 kt 8,110 30 kt mon-fired 15,000 kt 10,000 kt four FC3981 278,000 kt 8,110 30 kt mon-fired 11,500 kt 4,900 kt four FC3981 213,500 kt 10,000 kt 11,600 kt 11,600 kt 11,600 kt four FC3811 215,000 kt 11,600 kt 11,600 kt 11,600 kt 11,600 kt four Fc384 11 11 11,600 kt 11,600 kt 11,600 kt four Fc384 11 11 11,600 kt 11,600 kt four Fc384 four Fc384 11 11 11,600 kt 11,600 kt 11,600 kt four Fc384 11 11 11,1,600 kt <	313.000 kt 7,000 kt 5,000 kt 60 kt mon-fired 32,000 kt 57,000 kt fore PC391 239.000 kt 8,110 30 kt mon-fired 33,000 kt 33,000 kt fore PC391 239.000 kt 8,110 30 kt mon-fired 33,000 kt 33,000 kt fore PC391 239.000 kt 8,110 30 kt mon-fired 15,000 kt 0,000 kt fore PC391 239.000 kt 8,110 30 kt mon-fired 15,000 kt 15,000 kt fore PC3911 239.000 kt 8,110 30 kt mon-fired 15,000 kt 15,000 kt fore PC311 239.000 kt 8,110 30 kt mon-fired 21,500 kt 16,500 kt fore PC311 239.000 kt 8,100 kt 16,400 kt 12,500 kt 12,500 kt fore PC311 240.000 kt 8.110,400 kt 8.100 kt 12,500 kt 12,500 kt fore PC31 240.000 kt 9.100 kt 9.100 kt 12,500 kt 12,500 kt fore PC31 240.000 kt 7,500	0 104,000 km 8,050
332,000 kV 7,800 60 Nz mon-fired 220,000 kV 137,000 kV four FG381 138,000 kV 8,110 30 Nz mon-fired 81,000 kV 40,000 kV mon-fired 60,000 kV mon-fired 61,000 kV mon-fired 21,55,000 kV mon-fired 61,000 kV mon-fired 61,000 kV mon-fired 21,55,000 kV mon-fired 21,50,00 kV mon-fired 21,55,000 kV mon-fired 21,50,000 kV mon-f	332,000 kk 7,800 60 kk non-fired 220,000 kk 137,000 kk four FG381 108,000 kk 8,110 50 kk non-fired 29,000 kk 40,000 kk four FG381 278,000 kk 8,110 50 kk non-fired 15,000 kk 40,000 kk four FG381 278,000 kk 8,110 50 kk non-fired 155,000 kk 40,000 kk four FG381 278,000 kk 8,110 50 kk non-fired 155,000 kk 155,000 kk four FG381 278,000 kk 8,110 50 kk non-fired 21,5500 kk four FG381 278,000 kk 8,1,000 kk 10,000 kk 4,500 kk four France 5 271 20,60 kk 10,000 kk 10,135 kk one France 5 four France 6 271 272,500 kk 11,450 kk 11,450 kk 11,450 kk one France 6 271 272,500 kk 11,450 kk 11,450 kk four France 6 four France 6 four France 7 <t< td=""><td>33,000 ki 7,800 60 ki mon-fired 220,000 ki 137,000 ki four F63911 278,000 ki 8,110 30 ki mon-fired 81,000 ki 80,000 ki more F69111 278,000 ki 8,110 30 ki mon-fired 81,000 ki 80,000 ki more F69111 278,000 ki 8,110 30 ki mon-fired 81,000 ki 45,000 ki more F69111 278,000 ki 8,110 30 ki mon-fired 10,300 ki 125,000 ki more F69111 218,000 ki 8,110 30 ki mon-fired 13,300 ki 10,135 ki more F7386 118,000 ki 8,110 30 ki mon-fired 23,300 ki 11,450 ki more F7386 118,000 ki 8,130<ki< td=""> mon-fired 23,300 ki 11,450 ki more F7386 118,000 ki 8,110 90 ki mon-fired 23,300 ki 11,450 ki more F7386 118,000 ki 8,116 91,400 ki 114,500 ki 24,120 ki more F7386 118,000 ki 8,13,140 k</ki<></td><td>32,000 M 7,800 60 Hz con-fired 220,000 M 107,000 M four restant 218,000 W 8,210 30 Hz con-fired 29,000 M 8,000 W core 218,000 W 8,110 30 Hz con-fired 15,000 W 8,000 W res res111 218,000 W 8,110 30 Hz con-fired 15,000 W 8,100 W res res111 218,000 W 8,110 30 Hz con-fired 15,000 W 4,500 W res res111 218,000 W 8,110 30 Hz con-fired 15,000 W 4,500 W res res111 218,000 W 8,110 30 Hz con-fired 21,300 W res res111 218,000 W 11,450 W 11,450 W 11,450 W res ress 11 218,000 W 11,450 W 11,450 W 11,450 W res res ress 11 218,000 W 10,000 W 21,200 W 11,450 W res ress 14 218,000 W 11,450 W 11,450 W res ress 14 14,40 W 218,000 W 11,450 W</td><td>0 214,000 km 7,980</td></t<>	33,000 ki 7,800 60 ki mon-fired 220,000 ki 137,000 ki four F63911 278,000 ki 8,110 30 ki mon-fired 81,000 ki 80,000 ki more F69111 278,000 ki 8,110 30 ki mon-fired 81,000 ki 80,000 ki more F69111 278,000 ki 8,110 30 ki mon-fired 81,000 ki 45,000 ki more F69111 278,000 ki 8,110 30 ki mon-fired 10,300 ki 125,000 ki more F69111 218,000 ki 8,110 30 ki mon-fired 13,300 ki 10,135 ki more F7386 118,000 ki 8,110 30 ki mon-fired 23,300 ki 11,450 ki more F7386 118,000 ki 8,130 <ki< td=""> mon-fired 23,300 ki 11,450 ki more F7386 118,000 ki 8,110 90 ki mon-fired 23,300 ki 11,450 ki more F7386 118,000 ki 8,116 91,400 ki 114,500 ki 24,120 ki more F7386 118,000 ki 8,13,140 k</ki<>	32,000 M 7,800 60 Hz con-fired 220,000 M 107,000 M four restant 218,000 W 8,210 30 Hz con-fired 29,000 M 8,000 W core 218,000 W 8,110 30 Hz con-fired 15,000 W 8,000 W res res111 218,000 W 8,110 30 Hz con-fired 15,000 W 8,100 W res res111 218,000 W 8,110 30 Hz con-fired 15,000 W 4,500 W res res111 218,000 W 8,110 30 Hz con-fired 15,000 W 4,500 W res res111 218,000 W 8,110 30 Hz con-fired 21,300 W res res111 218,000 W 11,450 W 11,450 W 11,450 W res ress 11 218,000 W 11,450 W 11,450 W 11,450 W res res ress 11 218,000 W 10,000 W 21,200 W 11,450 W res ress 14 218,000 W 11,450 W 11,450 W res ress 14 14,40 W 218,000 W 11,450 W	0 214,000 km 7,980
138,000 kv 8,210 30 kt non-fired 81,000 kv 40,000 kv one F0311 278,000 kv 8,110 30 kt non-fired 15,000 kv 83,000 kv three F0311 418,000 kv 8,110 30 kt non-fired 15,000 kv 125,000 kv three F0311 418,000 kv 8,110 30 kt non-fired 248,000 kv 125,000 kv three F0311 418,000 kv 8,110 30 kv non-fired 21,530 kv 125,000 kv three F0311 418,000 kv 8,110 30 kv non-fired 21,530 kv 10,135 kv one Frame 3 418,000 kv 10,000 kv 10,000 kv 11,450 kv 11,450 kv one Frame 3 418,000 kv 11,450 kv 0,110 kv 0,1130 kv one Frame 3 418,000 kv 11,450 kv 11,450 kv one Frame 3 418,000 kv 11,400 kv 11,450 kv one Frame 3 418,000 kv 116,470 kv 00,110 kv one Frame 3 418,000 kv 116,470 kv 11,400 kv one Frame 3 419,400 kv 116,470 kv 11,400 kv one Frame 3 410,400 kv 116,470 kv 10,400 kv one Frame 3	13,000 kv 8,110 30 kt 6000 kv 40,000 kv	13,000 ku 6,100 30 kt 6000 ku 63,000 ku 63,000 ku 69,000 ku 60,000 ku	13,000 M 6,100 M 6,200 M 6,200 M 6,200 M res F0311 218,000 M 6,110 9 Rz constitued 15,000 M 5,000 M res F0311 218,000 M 8,110 9 Rz constitued 15,000 M 51,000 M res F1240 218,000 M 8,110 9 Rz constitued 15,000 M res F1240 218,000 M 8,100 9 Rz constitued 21,500 M res F1240 211 20 M 15,000 M 8,100 W 15,500 M res F1240 211 211 211 21,500 W 21,500 M res F1240 215,000 M 6,150 211 2110,000 M res F1240 215,000 M 6,150 211 21,500 M res F1240 215,000 M 7,23,40 M 21,240 M res F1240 res F1240 215,000 M 7,23,00 M 21,240 M res F1240 res F1240 res F1240 215,000 M 7,23,00 M 7,300 M 21,240 M res F1240 res F1240 <tr< td=""><td>0 323,000 ku 7,920</td></tr<>	0 323,000 ku 7,920
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278,000 kv 8,110 30 kt non-fired 165,000 kv 125,000 kv three F0311 418,000 kv 8,110 30 kt non-fired 248,000 kv 125,000 kv three F0311 30/60 kt non-fired 10,000 kv 1,500 kv non-fired 10,000 kv 1,500 kv non-fired 30/60 kt non-fired 21,550 kv 10,135 kv one Frame 3 30/60 kt non-fired 30,440 kv 11,450 kv one Frame 3 30 kt non-fired 31,500 kv 44,600 kv one Frame 3 30 kt non-fired 31,600 kv 44,600 kv one Frame 3 30 kt non-fired 31,600 kv 44,600 kv one Frame 3 50 kt non-fired 21,500 kv 44,600 kv one Frame 3 50 kt non-fired *16,470 kv 60,310 kv one Frame 3 50 kt non-fired *16,470 kv 10,00 kv one Frame 3	238,000 kv 8,110 30 kz non-fired 105,000 kv 33,000 kv 105,000 kv 105,000 kv 105,000 kv 106,000 kv 105,000 kv 106,000 kv 1	278,000 W 8,110 30 Hz non-fired 10,000 W 125,000 W three F0311 50/60 Hz non-fired 13,550 kW 10,115 kW one Frame 3 50/60 Hz non-fired 13,550 kW 10,115 kW one Frame 3 50/60 Hz non-fired 13,550 kW 10,115 kW one Frame 3 50/60 Hz non-fired 13,550 kW 10,115 kW one Frame 3 50 Hz non-fired 13,550 kW 10,115 kW one Frame 3 50 Hz non-fired 13,550 kW 11,450 kW one Frame 3 50 Hz non-fired *116,470 kW 60,310 kW one Frame 3 50 Hz non-fired *13,290 kW 10,41,240 kW one Frame 3 75,000 kW 6,130 50 Hz non-fired *13,240 kW fon 75,000 kW 73,500 kW 24,00 kW 24,00 kW fon 75,000 kW 73,500 kW 10,500 kW 10,500 kW fon 50 Hz non-fired *13,500 kW fon fon 125,000 kW 73,500 kW 13,500 kW 10,500 kW fon fon	Z78,000 W 8,110 30 Hz constitued 105,000 W 5,000 W trace rosall	0 123,000 ku 8,410
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	75,000 kW 8,150 50 Hz gas-fired 50,000 kW 25,000 kW two 55341 248,100 kW 7,850 50 Hz non-fired 169,800 kW 78,500 kW two 59001	75,000 kW 8,150 50 Hz gas-fired 50,000 kW 25,000 kW two 65341 248,100 kW 7,830 50 Hz non-fired 169,600 kW 78,500 kW two 69001 	75,000 ku 6.150 50 Hz gan-fired 50,000 ku 25,000 ku two 65141 246,100 ku 7,650 50 Hz non-fired 19,200 ku 78,500 ku two 69001 60 Hz non-fired 57,250 ku 21,900 ku two 67146-5 60 Hz non-fired 114,500 ku 48,450 ku two 67146- 111,700 ku 8,100 50 Hz non-fired 229,000 ku 2,500 ku effet 67146 50/60 Hz non-fired 25,400 ku 12,400 ku one 513	7 680,000 kW 7,586
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73,000 kV 8,130 50 Hz 643-fired 50,000 kV 25,000 kV two 65 kl 248,100 kV 7,850 kV two 6001 248,100 kV 78,500 kV two 6001	60 Hz non-fired 57,250 ku 23,900 ku two CFT4C-3		11,700 kW 8,100 30 Hz gas-fired 7,000 kW 4,700 kW one 519 	50 126, 580 ku 8, 135
73,000 kV 8.130 50 Hz 643-fired 50,000 kV 25,000 kV two C5141 246,100 kV 7,530 50 Hz non-fired 169,600 kV 78,500 kV two C6001 60 Hz non-fired 57,230 kV 23,900 kV two C714C- 60 Hz non-fired 114,500 kV 48,550 kV four C714C- 60 Hz non-fired 229,000 kV 97,560 kV effekt C714C	60 Hz non-fired 51,250 ku 23,900 ku two CFT4C-3 60 Hz non-fired 114,500 ku 48,450 ku four CFT4C- 60 Hz non-fired 229,000 ku 97,580 ku eight CFT4C		50/60 Hz non-fired 23,400 kM 12,400 kV one 01/mpus 50/60 Hz non-fired 20,600 kM 24,800 kV tvo 01/mpus	24 11, 700 km 8, 100
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TABLE 1.2.2.2.-1

PROJECTED OPERATING CHARACTERISTICS OF DEMONSTRATION ENGINES

						Temperati	ure	
	Time of	Output	Heat Rate Btu/kWh	Pressure Ratio	Flow 1b/sec	Turbine Inlet E	xhaust	Comments
urtis Wright	1988+	130 MM	9800 (est)	17	470	3000 1	300 (est)	Based on DOE High Temperature Turbine Technology Program - Aircooled (Transpiration) Vanes and Blades
General Electric	1988+	NH EL	10200 (est)	12.1	301	2600 1	273	Based on DOE High Temperature Turbine Technology Pro- gram - Water Cooled Vanes and Blades
шс	1988+	140 MM	10600 (est)	12	860	2400	1150	Based on Uprate of EPRI Reliable Engine Program - Aircooled (Convective) Vanes and Blades
Hest inghouse	1961	95.3 M	10800	14.2	194	2066 (est)	1026	50 Hz Version of W501
0101D/701D	1981	118.6 M	10800	14.2	910	2066 (est)	1024	Rased on a 50 Hz scaled-up version of 501

Curtis Wright

The technology path chosen by Curtis Wright uses the transpiration cooling by air of both blades and vanes to achieve operating temperatures of 2600 F and above.

A two-shaft, 17:1 pressure ratio engine with free turbine (Fig. 1.2.3.1-1), a scaled version of the C-W 6515 turbine was selected for the basic layout. Cooling air from the compressor discharge is used to cool the turbine blade and vanes (Fig. 1.2.3.1-2). The air foils (Fig. 1.2.3.1-3) consist of a structural spar with spanwise radial channels to which are electron beam welded porous metal skins. Cooling air from the hub is fed through each of the channels and then through the skin. The structural spars are kept to reasonable temperatures (1300 F - 1500 F) and the skin, which is non-load carrying, is protected by a film of relatively cool air.

Tests to date have demonstrated the capability to operate at 2600 F to 3000 F. These tests have been carried out using test cascades (not full engine configurations) for limited times. Only clean, liquid fuels have been used, although a blended fuel gas resembling low-Btu gas will be tested.

The results to date have been encouraging as far as the capability of attaining high temperature operation is concerned. Problems with over temperature on structural members has been resolved.

The combustor arrangement anticipated for use with the multi-spool engine is annular wrap-around combustor (Fig. 1.2.3.1-4). (An annular configuration resembles a torous. There are no individual cans, just a continuous section.) Cooling air keeps the wall temperatures to a 1600 F maximum level. No significant testing has been done on this combustor concept using low-Btu fuel.

General Electric

The technology path chosen by G.E. is the use of water to cool the vanes and blades to achieve operating temperature of 2600 F.

A single shaft 12.1:1 pressure ratio engine approximately the same size as the current Frame 6 was selected for the basic layout (Fig. 1.2.3.1-5). Cooling water is circulated in a closed loop for vane cooling. Water enters the vane from the bottom (Fig. 1.2.3.1-6) and exits at the top after making one more downward pass.



FIG. 1.2.3.1-1



COOLING CONFIGURATION FOR HIGH TEMPERATURE TURBINE



TRANSPIRATION AIR - COOLED TURBINE BLADE

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NOZZLE COOLING CIRCUITS

The airfoils are of two different constructions. The initial vane and blade are of composite construction (Fig. 1.2.3.1-7 shows the first-stage blade) using copper alloy inserts with nickel alloy skins. Coolant flows through nickel alloy tubes. The copper alloy inserts are supported by nickel alloy spars. This use of the copper is to assure even heat transfer through the blade/vane.

The blades and vanes for the second stage and beyond are of monolithic structure and are being considered as a backup for the composite first-stage vanes (Fig. 1.2.3.1-8). Water flows up the blade under the influence of a centrifugal field. Approximately 65 percent of the blade coolant is dumped overboard.

A sectoral combustor has been selected by G.E. (Fig. 1.2.3.1-9). This type of combustor is a compromise between the can-type and full annular combustor in that the torus is made up of a number of sectors, each with its own fuel system. In the G.E. system, the combustor is equipped with oil nozzle for startup and part load assistance.

General Electric has had good success with the water cooling. The vanes present no problems from the cooling view point although there have been structure problems. Water cooling of the blades has been demonstrated, but it appears that the monolithic structure is structural superior to the composite structure. Water purity requirements are high (deionized boiler feed water) and the water is lost overboard. None of these are unsurmountable problems.

1.2.3.2 High Reliability Gas Turbine Program

The High Reliability Gas Turbine Program (HRGT) was initiated by EPRI in 1978 with the objectives of defining those areas in gas turbine and combinedcycle design that have had poor records for reliability and identifying design changes to increase the system reliability to levels as least as good as steam turbine system. Three contractors, GE, United Technologies and Westinghouse, were selected. This program is an ongoing program, and no reports have been released by EPRI to date. Discussion centers on the UTC contribution.

The engine selected as the starting point of the HRGT program was the V84.3, the engine which would result from the UTC/KWU joint venture. This engine is a single shaft, two bearing engine with a pressure ratio of 12.1 (Fig. 1.2.3.2-1). In line with the EPRI philosophy of high reliability through extension of state-of-the-art technology, the turbine rotor inlet temperature of this machine would be 2120 F (combustor exit 2210 F) and the output will be 120 MW. Compressor discharge air is used to convectively cool the vanes and blades. The first stage vane cooling is shown in Fig. 1.2.3.2-2. The leading edge of the blade is cooled by air flowing radially in a channel formed by an

FIG. 1.2.3.1-7

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FIRST-STAGE COMPOSITE BLADE SECTION



No Mar



GENERAL ELECTRIC LOW-BTU GAS SECTORAL COMBUSTOR

1.2-66


881-955343-9





insert and the airfoil wall. Trip strips enhance the heat transfer. The trailing section of the blade is cooled on the suction side in the same manner except the flow is chordwise. The pressure side is cooled partially by impingement and partially by channel flow. There are no holes in the blade to plug up due to dirty fuels. The first stage blade is also convectively cooled by radially flowing air (Fig. 1.2.3.2-3).

Two types of combustors have been considered for this engine; the conventional (by U.S. standards) in-line can-type combustor and the off-base silo-type combustor. The silo-type combustor (Fig. 1.2.3.2-4), while initially more expensive offers advantages of high reliability, both for the combustor and for downstream elements and ease of tailoring for different fuels and for emission control.

All the elements of the HRGT program are either state-of-the-art or limited extensions thereof. These design features could be incorporated into engines which could appear in the 1984-85 time frame.

1.2.3.3 Potential Problem Areas

The major problem facing the use of gas turbines for base-load utility systems is a perceived lack of reliability. Thus, programs aimed at quantum jumps in technology may not achieve their goal of producing technology acceptable to the market. It must be understood that it is the market, and not technology, which will make the concept of advanced power systems a reality. Unless the industry accepts the product and furnishes enough money through purchases to amortize the hundreds of millions of dollars required to develop technology, the rate of progress will be very slow.

This problem is excebated by the fact that the quantum jumps in technology do not necessarily result in quantum jumps in performance. For example, Fig. 1.2.3.3-1 shows the performance for three levels of technology. The 2400 F and 3000 F performance levels differ by less than two percent; however, the 2400 F level of technology is only a modest extension of the state-of-the-art whereas the 2600 F/3000 F technology, as currently structured, requires quantum advances.

The combined -cycle systems using advanced technology gas turbines will compete against conventional steam stations with flue gas desulfurization (FGD) having performance levels of 35 percent or less, and against nuclear plants having performance of 32-33 percent. Thus, extracting the last bit of performance out of the system at the expense of very high R&D costs and potential reliability problems does not appear an attractive scenario.







TURBINE INLET TEMPERATURE EFFECT ON PERFORMANCE

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The path appearing most attractive would increase turbine temperatures through modest extensions of state-of-the-art techniques. Thus, one could envision air-cooled 2200 F turbines in the 1984-85 time frame followed by aircooled 2400 F turbines in the 1988 time frame and, ultimately, air-cooled 2600 F turbines in the 1995 time frame. The use of air cooling throughout would allow updating of already installed machines at relatively low cost. The substitution of a different cooling medium would increase any retrofit costs substantially.

1.2.4 Bibliography

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2.2 Description of Advanced Combined-Cycle Systems

The operating characteristics of advanced gas turbine systems and combinedcycle systems are described and, where pertinent, system design schemes presented. The baseline technology to be used in subsequent tasks is described.

2.2.1 Establish Operational Characteristics

The operational characteristics of commercially available and projected engines have been tabulated in Section 1.2. In addition to these characteristics, the requirements for fuel in terms of cleanliness will be presented and the environmental standards which currently have to be met by gas turbines will be discussed.

2.2.1.1 Gas Turbine/Combined Cycle Operational Characteristics

Based upon the results of Task 1.2, the ranges of probable operating characteristics for advanced gas turbine and combined cycle power plants are given in Table 2.2.1.1-1.

The operation of combined-cycle systems is routinely accomplished with no hazards beyond those associated with rotating machinery and high-temperature steam. Potential air pollution problems can be avoided by meeting the EPA standards for gas turbine operation described in Section 2.2.1.3.

2.2.1.2 Fuel Requirements

The gas turbine has often been touted as being able to run on nearly any fuel from peanut oil to blast furnace gas. While it is true that these fuels may be burned, the reliability and lifetime of the turbine are direct consequences of fuel characteristics.

Because of severe corrosion problems encountered when alkali metals, vanadium, sulfure, etc., are present in the fuel, there are strict requirements for fuel cleanliness. Erosion due to ash or other particulate carryover can also be a problem and thus, there is a limit on particulates.

Specifications for liquid and gaseous fuels are given in Tables 2.2.1.2-1 and -2, respectively. It must be noted that there are no actual specifications for coal-derived gaseous fuels. The values in Table 2.2.1.2-2 are based upon extrapolation of high Btu requirements or maximum allowable concentrations to meet pollution regulations.

2.2.1.3 Emission Standards for Gas Turbines

The emission standards for gas turbines are given in Table 2.2.1.3-1. Included for comparison are the steam station regulations. Combined-cycle plants must meet

TABLE 2.2.1.1-1

RANGE OF OPERATING CHARACTERISTICS OF ADVANCED GAS TURBINE AND COMBINED CYCLE SYSTEMS

Gas Turbine

Output, MW	75 - 140
Flow rate, 1b/sec	300 - 900
Pressure ratio	12 - 17
Turbine Inlet Temp, F	2200-3000
Metal Temp., F	1100 - 1600
Exhaust Temp., F	1000 - 1150
Heat Rate, Etu/kWh	9500-10500

Steam Turbine

Output, MW Pressure, psi Temp., F Type

Combined Cycle

Output, Mw Heat Rate, Btu/kWh

100 - 400(1)
$1000 - 2400^{(2)}$
350 - 1000
Reheat multi-

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300 - 1000(3)6200 - 7500(4)

pressure

- Not limited by technology. One to four gas turbines will provide waste heat to generate steam for one steam turbine.
- (2) Pressure level fixed by optimizing overall system.
- (3) Multiple engines and one steam turbine.
- (4) Based on clean fuel delivery to gas turbine.

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TABLE 2.2.1.2-1

LIQUID FUEL SPECIFICATIONS

Property	Distillate ⁽¹⁾	Heavy 011(2)
Net Heating Value, Btu/1b	18400	18400
Sulfur, wt %	1(3)	1(3)
Metal Contaminants, ppm		
Vanadium	0.5	0.5
Sodium plus Potassium	1.0	1.0
Calcium	2.0	2.0
Lead	1.0	1.0
Initial Ash Sticking Pt., F	1652	1652
Ash, ppm	50	50
Carbon-Hydrogen Ratio	No Limit ⁽⁴⁾	No Limit ⁽⁴⁾

- All JP-type fuels, diesel 1 and GT-1, diesel 2 and GT-2, GT-3 and distillate-type cuts of coal derived liquid fuels.
- (2) GT-3, residual No. 5 and 6, Bunker C and heavy cracks or heavy oil-type cuts of coal derived liquid fuels.
- (3) Air pollution requirements reduce this to a maximum of 0.8% S content.
- (4) The lowest C/H ratio is recommended.

TABLE 2.2.1.2-2

GASEOUS FUEL SPECIFICATIONS

Property	High Btu(1)	Low Btu ⁽²⁾
Heating Value, Btu/ft ³	950 - 1200	100350(3)
Sulfur	1.8 mol. % ⁽⁴⁾	< 0.075 mol % or less than amount required

to form 0.6 ppm alkali metal sulfates(5)

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Metals

Vanadium	< 0.2 ppm (wt)	<0.03 ppm
Sodium plus	< 0.6 ppm (wt)	See sulfur spec.
Calcium	< 0.1 ppm (wt)	< 0.012 (wt)
Lead	< 0.1 ppm (wt)	< 0.012 (wt)
Copper	<0.2 ppm (wt)	< 0.025 (wt)
Particulates	<0.08 1b/10 ⁶ ft ³	$0-3\mu$ < 0.025 gr/ft ³
		4-54 <0.03 gr/ft ³
		$>5\mu$ < 0.0036 gr/ft ³

(1) P&WA Specification 527

- (2) There are no definitive specifications for low-Btu fuel gas. The values listed are suggested values based upon extrapolation from natural gas or other considerations.
- (3) Fuel gas with heating values less than 100 Btu/ft³ and greater than 350 Btu/ft³ have been successfully combusted. The range listed encompasses the majority of coal-delivered fuels.

(4) Does not meet pollution requirement of 0.45 mol %.

(5) Based on typical 150 Btu/ft³ fuel.

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TABLE 2.2.1.3-1

EMISSIONS REGULATIONS

Pollutant	Gas Turbine	Coal-Fired Steam
Sulfur	0.015 mol % SO ₂ (0.9 1b SO ₂ /10 ⁵ Btu) or 0.8% (wt) S in fuel	1.2 1b SO ₂ /10 ⁶ Btu max with 90% reduction over uncontrolled level
Nitrogen Oxides	0.0075 to 0.0125 mol % 0.31 to 0.52 1b/10 ⁶	0.5 lb/10 ⁶ Btu for sub-bituminous or coal-derived fuels; 0.6 lb/10 ⁶ Btu for other coals
Particulates	No standard	0.03 1b/106 Btu

the gas turbine regulations unless the steam portion is supplemently fired. Then the gas turbine must meet the turbine specifications while the steam portion must be treated as a fired system and meet the required regulations for oil- or gas-fired steam stations.

2.2.2 Summarize System/Subsystem Design

The gas turbine-based combined-cycle power system is shown in schematic form in Fig. 2.2.2-1. A range of operating parameters is included in Fig. 2.2.2-1 corresponding to the values given in Table 2.2.1.1-1.

Subsystem drawings for the gas turbines have been shown in Section 1.2.3, while a front view giving combustor details is shown in Fig. 2.2.2-2. A subsystem drawing of the steam bottoming cycle is shown in Fig. 2.2.2-3 while details of the steam turbine are shown in Fig. 2.2.2-4.

2.2.3 Describe System Baseline Technology

The data presented in Tasks 1 and 2 on gas turbines and combined-cycle systems have a wide range from which to select a baseline technology. There are, however, guide lines for selection which can be used. These are:

- . date of appearance
- . likelihood of technology attainment
- . acceptance by utility market place

All of these are interrelated and are, unfortunately, subjective in nature.

The advanced gas turbines are being developed, for the most part, as power systems for use with coal-derived fuels. While there will continue to be a market for petroleum-fired systems in several parts of the world, the major manufacturing nations are anticipating the need for synthetic fuels. At this time, only one demonstration combined-cycle plant using coal-derived fuels is in operation although several more are planned. Thus, it appears that it will be the end of the 1980 decade before there will be a market for these engines.

With a target date of 1988, it is possible that the goals of the HTTTP will be met, i.e., turbine inlet temperatures of 2600 F or higher will have been demonstrated. Given, however, the state of the present government and industry R&D funding and the state of the marketplace, it is improbable that such technologies will be deemed commercial. This means that there could be demonstration plants (or units) operating at these advanced levels but no deliveries of equipment with warranted performance.

According to the R&D path being followed by EPRI, the most important requirement for utility acceptance of gas turbine is reliability. The utility industry

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TYPICAL WASTE-HEAT COMBINED GAS AND STEAM TURBINE SYSTEM





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TYPICAL STEAM TURBINE GENERATOR

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appears convinced that combined-cycle power plants could offer superior performance compared to conventional alternatives. Their widespread acceptance, however, even at modest technology levels is predicated on the demonstration of reliability similar to that of alternative plants.

With the foregoing discussions in mind, it would appear that the selection of technology for the gas turbine should be a compromise between the HTTTP levels and the EPRI HRGT levels. This would be an engine using technology which is a logical progression from that in the HRGT towards that possible in the HTTTP. Based on this, an uprated version of the UTC engine proposed for the EPRI HRGT program has been projected. The operating characteristics of this engine are:

Flow rate - 860 lb/sec Pressure ratio - 12 Turbine inlet - 2400 F Power output - 140 MW (nominal)

To achieve these operating characteristics will require an increase in the turbine cooling effectiveness previously discussed in Section 1.2.3.2. It will also require a low-emission combustor to meet stringent NO_x emission levels.

2.2.3.1 Compressor

The compressor will consist of 16 stages giving an overall pressure ratio of 12:1. Variable inlet guide vanes will be used to allow good start up and part-load operation.

2.2.3.2 Combustor

The silo-type combustor will be utilized. In order to meet the requirements for $low-NO_x$ emissions, the combustor head will be redesigned to use the rich-lean burn concept. This concept meets NO_x regulation without the use of water injection. As applied to gaseous fuels, the concept is shown in Fig. 2.2.3-1. Because of the large size of the silo-burner, it is relatively easy to achieve a tailored combustion pattern.

2.2.3.3 Turbine

A four-stage turbine would be used. The major change required is in the type of blade/vane cooling. Rather than the monolithic blade with internal cooling passages, a bonded blade would be used (Fig. 2.2.3-2). The use of the multi-piece blade will allow much closer control over coolant channel dimensions and locations, thus allowing metal temperatures to remain at the desired level at higher firing temperatures but with no appreciable increase in cooling flow.





2.2.3.4 Combined Cycle System

A schematic of the baseline combined-cycle system is shown in Fig. 2.2.3-3. The operating characteristics and estimated performance are given in this figure. A two engine module has been selected for presentation. Any multiple of engines may be used. Each would have its own heat recovery steam generator feeding into a central steam turbine.

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OVERALL PLANT SYSTEM CONCEPT

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3.2 Evaluation of Coal-Derived Fuels for Advanced Combined-Cyle Power Systems

The objective of this task is to determine the overall power plant efficiency of a combined gas turbine and steam turbine (combined-cycle) system using medium-Btu gas delivered to the power plant via pipeline. The properties of the gas delivered to the power plant will be obtained and any further processing, if necessary, will be identified. Based upon the fuel gas characteristics, the performance of the combined cycle power plant will be described.

3.2.1 Definition of Fuel Gas Requirements

A description of the suggested fuel gas properties for coal-derived gases was prepared as part of Task 2.2.1 and presented in Table 2.2.1.2-2. As mentioned in the description, the allowable quantities of alkali metals and other heavy metals must be quite low. This should not pose a problem, because these low levels are attained during the low-temperature clearup associated with sulfur removal (Refs. 3.2.-1 and -2.).

The oxygen-blown Texaco and British Gas Corp. (BGC) slagging gasifiers were selected as the gasifiers to provide medium-Btu fuel gas for transport in pipelines to the power generation site. Properties of the fuel gas from each of these gasifiers, based upon Ref. 3.2.-3, are given in Table 3.2.1-1. The total sulfur content of 100 ppm (H_2S + COS) is that agreed upon by NASA and the Contractor (Ref. 3.2-4) as being typical of pipeline quality fuel gas.

The combustor pressure is 175 psia so that the fuel delivery pressure at the power plant boundary must be 175 psia plus that required for the fuel control system. For medium-Btu fuel gases, an overall pressure drop of 75 psi between the power plant boundary and the combustor is adequate. Thus, a pressure of at least 250 psia would be required at delivery. It is anticipated that the Texaco and BGC gasifiers would be operated at pressures well beyond this value.

Fuel delivery temperature is not critical unless it is low enough to allow condensation. For the moisture content given in Table 3.2.1-1, temperatures should be above 75 F. An average temperature of 86 F has been assumed. An upper limit on temperature is set by the valves and seals in the fuel controls and is well beyond any value typically associated with pipeline delivery of gas.

The required flow rate of fuel gas is a function of heating value and engine power levels. For a single engine at design point operations (nominally 150 MW), the flow rate for gas from the Texaco gasifier would be 309,850 lb/hr while that from the BGC would be 215,550 lb/hr.

3.2.2 Identification of Power Generation Facility

The combined-cycle power system has been described in Section 2.2.3. The use of medium-Btu fuel gas does not require any additional equipment although individual equipment will vary in size and output depending on the fuel. A general schematic for the combined cycle power plant is given in Fig. 3.2.2-1. In brief, the operation is as follows:

Air (Steam 1 in Fig. 3.2.2-1) is compressed to approximately 175 psia in the gas turbine compressor. Upon exiting the compressor, a small fraction of the air is sent to the turbine for turbine vane/blade cooling while the remainder is used for combustion. The combustion products are expanded and the exhaust steam (3) is split into two streams and used to superheat steam (4) and to reheat steam (5). The exhaust streams are rejoined (6) in the boiler, pass through an economizer, the boiler for deaeration steam, and finally exit through the stack (9).

Condensate (10) is pumped to a low-pressure deaerator, and then pumped to boiler pressure before entering the economizer (16). From the economizer, the high-pressure water enters the boiler where it is changed to saturated steam (18). The steam is superheated, expanded to an intermediate pressure (20) and then reheated (21) prior to complete expansion in the intermediate- and low-pressure turbines (22).

The flow conditions (heat and mass) vary for each type of fuel gas. A heat and mass balance for the Texaco fuel gas system, based on one engine, is given in Table 3.2.2-1 while that for the BGC system is given in Table 3.2.2.-2. Material balances corresponding to these are given in Table 3.2.2-3 and 3.6.2-4 for the Texaco and BGC systems, respectively.

Because of the modularity of the combined-cycle system, an actual power plant could consist of a number of gas and steam turbines. However, the basic building block would be two gas turbines with two waste heat boilers generating steam for one steam turbine. The nominal output of the module would be 450 MW.

The estimated performance for this basic module for both fuel gas types is given in Table 3.2.2-5. The estimated performance of a distillate-fueled system is given for comparisons. The performance levels for all three fuels are comparable at approximately 50 percent. The differences are mainly due to the differences in mass flow rate through the engines. The fuel flow for the distillate fuel represents approximately 2.5 percent of inlet flow; for the Texaco, 10 percent; and for the BGC, 7 percent.

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TABLE 3.2.1-1

GAS COMPOSITIONS

	Texaco	Gas	BGC	Gas
Element	Moles/hr	Vol. %	Moles/hr	Vol. %
CH ₄	15.0	0.1	935	8.4
H ₂	5394	35.6	3414	30.6
CO	7940	52.4	6515	58.5
co ₂	1629	10.8	217	1.9
H ₂ S				
+ cos	n (n. 1 . 1997)	100 ppm	1997 - 199 3 - 1997 - 1997	100 ppm
N ₂	124	0.8	50	0.5
Ar	22	0.2	<1	0
H ₂ 0	15	0.1	11	0.1
	15139	100.0	11142	100.0

Heating Values, LHV

LHV,	Btu/scf	267.5	349.1
LHV,	Btu/1b	4949	6825

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TABLE 3.2.2-1

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HEAT AND MASS BALANCE FOR COMBINED CYCLE POWER SYSTEM USING TEXACO COAL-DERIVED FUEL

Station Number	Temperature (°F)	Pressure (Psia)	Flow (1b/sec) ⁽¹⁾
1	59	14.7	860.0
2	100	175.0	86.1
3	1191	14.8	946.1
4	1191	14.8	620.0
5	1191	14.8	326.1
6	907	14.8	946.1
7	716	14.8	946.1
ŏ	447	14.8	946.1
y	364	14.8	946.1
10	101	1.2	132.4
11	101	30.0	132.4
12	101	30.0	114.3
13	101	30.0	18.1
14	101	30.0	18.1
15	250	30.0	132.4
16	255	2844.0	132.4
17	606	2474.0	132.4
18	66 6	2474.0	132.4
19	1000	2400.0	132.4
20	656	600.0	132.4
21	1006	540.0	132.4
22	108	1.2	132.4

(1) All flow values are for one engine.

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TABLE 3.2.2-2

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HEAT AND MASS BALANCE FOR COMBINED CYCLE POWER SYSTEM USING BGC COAL-DERIVED FUEL

Station Number	Temperature (*F)	Pressure (Psia)	Flow (1b/sec)(1)
1	59	14.7	860.0
2	100	175.0	59.9
3	1179	14.8	919.9
4	1179	14.8	603.5
5	1179	14.8	316.4
6	902	14.8	919.9
7	716	14.8	919.9
8	454	14.8	919.9
9	373	14.8	919.9
10	101	1.2	125.3
11	101	30.0	125.3
12	101	30.0	108.1
13	101	30.0	17.2
14	101	30.0	17.2
15	250	30.0	125.3
16	255	2844.0	125.3
17	60.0	2474.0	125.3
16	666	2474.0	125.3
19	1000	2400.0	125.3
20	657	600.0	125.3
21	1000	540.0	125.3
22	106	1.2	132.4

(1) All flow values are for one engine.

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				,			TABLE	3.2.2-3						, , , , , , ,
, •						MATER PUWER SYS	RIAL BALAN STEM USING	CE FOR CO : TEXACO C	MBLNED CY OAL-DERLV	CLE ED FUEL				
	Statio	n Number	•	1		2	e		4		Ś			.0
	Comp.	М. И.	Lb/Hr ⁽¹⁾	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	l.b/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr
- ++	СН¢	16.04		ı	241	15.0		ı		ı		I		•
	H2	2.016		ı	10874	5394		ţ		I		ı		I
	8	28.01		ı	222394	1940		I		ł		ł		ı
-	C02	10.44		ı	71692	. 1629	421792	9584	276427	6281	145365	3303	421792	9584
-	H2S	34.08		I	I	ı		1		ı		8		ı
<u> </u>	SOS	60.07		ı	ı	I		I		I		t		ľ
5	22	32.00	713312	22291	ı	ł	499008	15594	327040	10220	171968	5374	499008	15594
4	4 2	28.02	2321933	82867	3474	124	2325403	82991	1523980	54 389	801400	28601	2325380	82990
٠.	_	39.94	41578	1041	879	22	42456	1063	27838	697	14618	366	42456	1063
æ	1 ₂ 0	18.02	19534	1084	270	15	117544	6523	11011	4274	40509	2248	117526	6522
2	lot al		3096357	107283	309829	15139	3406208	115755	2232302	75861	1173860	39892	3406162	115753
~	IV (1	ll value:	s based on	one engine										

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MATERIAL BALANCE FOR COMBINED CYCLE POWER SYSTEM USING TEXACO COAL-DERIVED FUEL

Stat i	n Number	F	7	~	20	6		10	-	11			12
Comp.	М.К.	Lb/Hr ⁽¹⁾	Mols/Hr	1.b/нг	Mols/Hr	Lh/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr
CH4	16.04												
H2	2.016												
8	28.01												
c02	10.44												
H2S	34.08												
COS	60.07												
U2	32.00					•							
N2	28.02												
<	39.94												
н ₂ 0	18.02							476737	26456	476737	26456	411469	22834
Total		3406162	115753	3406162	115753	3406162	115753	476737	26456	476737	26456	41 1469	22834

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						TABLE 3.2	2-3 (Cont	('P,				 	1
					MATE! POWER SY!	KIAL BALAN STEM USIN	NCE FOR COI 5 TEXACO CO	MBINED CYC Dal-derive	LE D PUEL				
Statio	radmun na	Ŀ	13	14		15		16		17		81	
Comp.	N. K.	Lb/нг ⁽¹⁾	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	·Lb/Hr	Mols/Nr
СН4	16.04												
H ₂	2.016												
8	28.01												
α ²	10° 44												
H2S	34.08												
SOS	60.07												
02	32.00												
N2	28.02												
<	39.94												
HZO	18.02	65268	3622	65268	3622	476737	26456	476737	26456	476737	26456	476737	26456
Totel		65268	3622	65268	3622	476737	26456	476737	26456	476737	26456	476737	26456
3.2-9													

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TABLE 3.2.2-3 (Cont'd.)

MATERIAL BALANCE FOK COMBINED CYCLE POWER SYSTEM USING TEXACO COAL-DERIVED FUEL

at io	n Number	L	61	N	0	21		22	
	Μ.Υ.	Lb/нг ⁽¹⁾	Mols/Hr	Lh/Hr	Mols/Hr	Lb/Hr	Mols/Hr	гр/нг	Mols/lir
	16.04								
	2.016								
	28.01								
	44.01								
	34.08								
	60.07								
	32.00								
	28.02								
	39 44								
	18.02	476737	26456	476737	26456	476137	26456	476737	26456
		476737	26456	476737	264 56	476737	26456	476737	26456

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						TABL	E 3.2.2-4						
					MATE POWER	RIAL BALAI System US	NCE FOR C	MBINED CI ML-DÉRIYI	rcle ed fuel				
St <i>a</i> t i	on Numbe	L	-		2	ũ		4		ď			
Comp.	M. K.	Lb/Hr ⁽¹⁾	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	1.b/Hr	Mols/Hr	Lb/Hr	Mole/H-		
ਝੈ	16.04			14997	506							i i i i i i i i i i i i i i i i i i i	111/8/04
н ₂	2.016			6883	3414								
8	28.01			182485	6159								
8 8	10.44			9550	217	337469	7668	221370	5030	116098	96 J.C	07766	
H2S	34.08				ı		I				2		900/
cos	60.03				I		ł						
0 ²	32.00	218612	16222		ľ	494624	15457	324480	10140	170144	2123	4 9 4674	16463
Zhi	28.02	2321933	82867	1401	50	2323334	82917	1524148	24 395	799186	2852	4666666	
<	39.94	41578	1041		ı	41578	1041	27279	683	14299	356	41578	1470
H2O	18.02	MESS1	1084	198		114932	6378	15396	4184	39536	2194	114932	6378
Tot al		3096357	107283	215514	11 142	3311937	113461	2172673	74432	1139263	39029	331 1937	113461

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(1) All values based on one engine.

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k61-9	5343-9		tols/Hr										21610	21610			
		12	Lb/Hr										389412	389412			
			ols/Hr										25038	25038			
		11	Lb/Hr M										451185	451185			
	E Fuel		ols/Hr										25038	25038			
~	-DERIVED	10	Lb/Hr M										451185	451185			
(Cont'd.	FOR COMB. BGC COAL		ols/Hr											74432			
E 3.2.2-4	BALANCE Em Using	6	b/Hr Mo											72673			
TABLI	MATERIAL Ower syst		ls/Hr l.											4432 21			
	2	30	о/Нг Мо											2673 7			
			1										1	2 2175			
		2	Mols/H											7443			
			Lb/Hr ^(I')											2172673			
		n Number	Μ.Ψ.	16.04	2.016	28.01	44.01	34.08	60.07	32.00	28.02	39.94	18.02				
		Statio	Comp.	СН ₄	H ₂	8	c0 ₂	H ₂ S	cos	02	N2	×	H2O	Total	3.2-	12	

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•			t at ion	.duo	H4	2	0	2 ^C	ss	SC	2	2		0 ²	ot a l
) Number	М. И.	16.04	2.016	28.01	44.0:	34.08	60.07	32.00	28.02	39.9k	18.02	
			يت مد بر	Lb/Hr ⁽¹⁾										61773	61773
			e	Mols/Hr										3428	34 28
			14	Lb/Hr										61773	61773
-	TAI	MATER POWER S		Mols/lir										3428	3428
•	BLE 3.2.2-	IAL BALAN(YSTEM USI	15	Lb/Hr										451185	451185
•	-4 (Cont'd	ce for cor Ng bgc cor		Mols/Hr										25038	25038
•	1.)	4BINED CY(AL-DERIVEI	16	Lb/Hr										451185	451185
•		CLE D FUEL		Mols/Hr										25038	25038
			17	Lb/нг										451185	451185
			•	Mols/Hr										25038	25038
			18	Lb/Нг										451185	451185
	VQT-3]]	343-3		Mc1s/Hz										2503	2503

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MATERIAL BALANCE FOR COMBINED CYCLE POWER SYSTEM USING BGC COAL-DERIVED FUEL

Statim Number 19 20 21 22 Comp. M.W. Lb/Hr ⁽¹⁾ Mols/Hr Lb/Hr Nols/Hr Lb/Hr Nols/Hr Lb/Hr Mols/Hr Mols/Hr		s/Hr										5038	8600
Station lumber 19 20 21 $Comp.$ $M.W.$ $Lb/Hr(1)$ $Mols/Hr$ Lb/Hr $Mols/Hr$	22	b/Hr Mole										451185 2	451185 2
Station Number 19 20 21 Comp. N.W. Lb/Hr(1) Mols/Hr Lb/Hr Lb/Hr Lb/Hr Cump. N.W. Lb/Hr(1) Mols/Hr Lb/Hr Lb/Hr Lb/Hr Cump. N.W. Lb/Hr(1) Mols/Hr Lb/Hr Lb/Hr Lb/Hr Cump. 16.04 N.W. Lb/Hr Nols/Hr Lb/Hr Lb/Hr H2 2.016 A 2.016 A A A A A U2 28.01 A A A A A A A U2 34.08 A A A A A A A V2 32.00 A A A A A A N2 28.02 A A A A A A N2 28.02 A A A A A A N2 28.02 A A A A A A N2 B A A		Mols/Hr l										25038	25038
Station Number 19 20 Comp. M.W. Lb/Hr ⁴ (1) Mols/Hr Mols/Hr Ch4 16.04 Lb/Hr ⁴ (1) Mols/Hr Mols/Hr H2 2.016 \times \times \times \times U2 28.01 \times \times \times \times U2 24.01 \times \times \times \times \times U2 24.01 \times \times \times \times \times \times U2 24.03 \times \times \times \times \times \times \times U2 24.08 \times	21	Lb/Hr										451185	451185
Station Number 19 20 Comp. N.W. Lb/Hr ⁽¹⁾ Mols/Hr Lb/Hr 1 U 16.04 $1b/Hr^{(1)}$ Mols/Hr Lb/Hr 1 H ₂ 2.016 1.04 1.04 1.04 1.04 U 2.016 44.01 1.04 1.04 1.04 U 28.01 1.04 1.04 1.04 1.04 U 28.01 1.04 1.04 1.04 1.04 1.04 H ₂ S 34.08 1.04 1.02 <		Mols/Hr										25038	25038
Station Number 19 Station Number 19 Comp. N.W. Lb/Hr ⁽¹⁾ Mols/Hr Ul_4 16.04 Mols/Hr Mols/Hr U_2 2.016 A A U_2 28.01 A A U_2 34.08 A A M_2 34.08 A A U_2 34.08 A A M_2 34.03 A A M_2 34.03 A A M_3 34.03 A A	20	Lb/Hr										451185	451185
Station Number 19 Station Number 10.04 Comp. M.W. Lb/Hr ⁽¹⁾ CH4 16.04 H2 2.016 CO 28.01 M2 34.08 N2 32.00 N2 28.02 N2 39.94 H2O 18.02 H2O 18.02 H2O 18.02 H2O 18.02 H2O 18.02 H2O 18.02		Mols/Hr										25038	25038
Station Number Comp. M.W. Comp. M.W. CH4 16.0% H2 2.016 CO 28.01 H2S 34.08 H2S 34.08 M2 32.00 N2 28.02 M2 32.00 M2 39.94 M2 39.94 M2 18.02 H20 18.02 H20 18.02 H20 18.02 H20 18.02 H20 18.02 H20 18.02	61	_{Lb/нг} (1)										451185	451185
Station Comp. Comp. CH4 H2 CO2 CO2 CO2 CO2 CO3 N2 N2 N2 N2 N2 N2 N2 N2 N2 N2 N2 N2 N2	Number	М. W.	16.04	2.016	28.01	44.01	34.08	60.07	32.00	28.02	39.44	18.02	
	Station	Comp.	СН ₄	H2	8	c02	Η2S	cos	02	N2	۷	Н ₂ 0	Total

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TABLE 3.2.2-5

PERFORMANCE ESTIMATES FOR COMBINED-CYCLE SYSTEMS

Fuel Type:	Distillate	<u>Coal</u> D	erived
	No. 2 Oil	Texaco	BGC
Gas Turbine:			
Flow Rate, 1b/sec	860	860	860
Pressure Ratio	12	12	12
Combustor Exit Temp., ^O F	2400	2400	2400
Exhaust Temp., ^O F	1162	1192	117 9
Stack Temp., ^O F	387	364	374
Turbine Area, in ²	483	521	506
Fuel System:			
Flow Rate, 1b/hr/engine	74758	309850	215550
Heating Value, Btu/1b (LHV)	18476	4948	6826
Combined Cycle Performance*:			
Gas Turbine Output, MW (2 engines)	271.4	306.8	292.8
Steam Turbine Output, MW (1 turbine) 143.6	165.4	156.2
Utility Losses, MW (per engine)	6.2	7.1	6.7
Total Net Power, MW	402.2	458	435.6
Heat Rate, Btu/kWh (LHV)	6865	6696	6755
Efficiency, %	49.7	51.0	50.5

* Steam System: Single pressure, 2400 psia/1000°F/1000°F Condenser Pressure = 2.5 in.

The additional flow rate due to the large amount of medium-Btu fuel could require modifications to the gas turbine to ensure that no mismatch between the turbine and compressor occurs. Usually, if flow area differences are less than 10 percent of the design basis (the distillate case), the changes can be accommodated by adjustment of turbine components. Larger differences could require redesign of the turbine. The turbine areas listed in Table 3.1.2-5 indicate that no redesign would be necessary.

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3.2.3 References

- 3.2-1 Robson, F. L., W. A. Blecher and V. A. May: Fuel Gas Environmental Impact: EPA 600/7-78-088, June 1978.
- 3.2-1 Blecher, W. A. and F. L. Robson: Assessment of Fuel Gas Cleanup Systems. Contract DE-AC21-79MC12050 with METC, November 1980.
- 3.2-3 Fluor Engineers and Constructors: An Economic Comparison of Molten Carbonate Fuel Cells and Gas Turbines in Coal Gasification-Based Power Plants. EPRI AP-1543, Sept. 1980.
- 3.2-4 Interim Project Briefing, NASA-MSFC, Huntsville, Alabama, December 17/18 1980.

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4.2 Evaluation of Integrated Gasification Combined-Cycle Power Systems

The objective of this task is to determine the overall power plant efficiency of an integrated gasification combined-cycle (IGCC) power plant. Two types of gasifiers will be considered and the performance of the two resulting IGCC power plants will be compared. Flow sheets for the gasifiers, the oxygenblown Texaco and British Gas Corp. slagging types, are those previously developed by the contractor as agreed upon by NASA and the contractor (Ref. 4.2-1).

4.2.1 Identification of IGCC Power Plant with Texaco Gasifier

The combined-cycle power system identified in Section 2.2.3 will be used as the basis of the IGCC power plant. The operating conditions are the same but the equipment output differs because of differing steam loads in the IGCC power plant. A schematic for the Texaco-based IGCC is given in Fig. 4.2.1-1.

In brief, the operation is as follows. Similar to the conventional combinedcycle plant of Section 2.2.3, air (Steam 1 in Fig. 4.2.2-1) is compressed to approximately 175 psia in the gas turbine compressor. Upon exiting the compressor, a small fraction of the air is sent to the turbine for turbine vane/blade cooling while the remainder is used for combustion. The combustion products are expanded and the exhaust stream (30) is split into two streams and used to superheat steam (31) and to reheat steam (32). The exhaust streams are rejoined (33) in the boiler. They then pass through economizers and the deaerator finally exiting through the stack (37).

Both the steam system and the fuel system differ from the conventional plant. Water from the condenser (38) is pumped to a low pressure and split into two streams. Stream 40 is further pumped and heated against the fuel gas stream (Q_1) to furnish process steam for use in the cleanup system. The other condensate stream (43) is heated against the fuel gas stream, then used to degreate the remaining flow prior to final pumping to boiler inlet pressure before the fist of two economizers (47). After this economizer, the stream is again split with a small stream being used to raise LP steam for process use while the majority passes through the second economizer (54). This stream is then split, some going to the waste heat boiler behind the gas turbine (55) and some to the heat recovery boiler in the fuel gas stream (56). These streams rejoin, pass through the superheater and are expanded in the high-pressure steam turbine. After reheating the steam is sent to the intermediate pressure turbine (67) with a side stream being separated to drive the air and oxygen compressors in the air separation plant (66). After full expansion in the low-pressure turbine, the steam enters the condenser (70).

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The fuel gas starts as a coal-water slurry (5) and oxygen (4) feeds to the gasifier. After gasification, the fuel gas stream is split, one stream being water quenched (13) to remove slag while the other passes through a boiler (8), then a fuel gas regenerator. Prior to going through the boiler, the hot gas temperature is moderated by a recycle stream of cool gas (18). The fuel gas is subsequently cooled, sulfur removed and then resaturated and reheated prior to expansion in a turboexpander (25) down to combustor pressure.

The heat and mass balance for the Texaco-based IGCC is given in Table 4.2.1-1 (flow is for one engine). Material balances are given in Table 4.2.1-2 (again based on one engine)

Because of the modularity of the combined-cycle system, an actual power plant could consist of a number of gas and steam turbines. However, the basic building block would be two gas turbines with two waste heat boilers generating steam for one steam tubine. The nominal output of the module would be 475 MW.

The coal gasification system is costly and benefits from scale and replication, thus, it is probable that the 475-MW module would be the minimum size plant. A more typical plant would probably consist of two such modules, i.e., four gas tubines and waste heat boilers feeding one steam turbine. Such a system would require ten gasifiers (two spares). The estimated performance for such an IGCC plant is given in Table 4.2.1-3.

4.2.2 Identification of IGCC Power Plant with British Gas Corporation Gasifier

The IGCC power plant using the BGC gasifier is similar to that using the Texaco gasifiers (Fig. 4.2.2-1 shows a schematic) except for variations in the steam raising sections.

Because of the lower exit temperature from the gasifier and the need to remove tars and solid carryover, there is no waste heat recovery boiler in the gasifier exit stream (stream 41 - Fig. 4.2.2-1). Rather, the heat removed from this stream prior to desulfurization is used to resaturate the fuel gas after cleanup. All the steam for power generation and process use is raised in the gas turbine exhaust, with the exception of a small amount of process steam raised in the gasifier jacket.

A heat and mass balance for the IGCC powerplant with the BGC gasifier is given in Table 4.2.2-1 and a material balance is given in Table 4.2.2-2. The performance of this system is given in Table 4.2.1-3.

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TABLE 4.2.1-1

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HEAT AND MASS BALANCE FOR GAS TURBINE COMBINED CYCLE POWER PLANT USING TEXACO OXYGEN-BLOWN GASIFICATION SYSTEM

Station Number	<u>Temperature (°F)</u>	<u>Pressure (psia</u>)	Flow (1b/sec) (1)
1	59	14.7	860.0
2	655	175	723.5
3	60	17	39.3
4	300	588	39.3
5	140	588	70.2
6	2400	588	102.2
7	2400	588	92.0
8	1850	588	128.7
9	1200	567	128.7
10	729	567	128.7
11	515	567	128.7
12	407	567	128.7
13	2400	588	10.2
14	400	567	14.4
15	406	567	143.1
16	352	567	146.8
17	352	567	36.7
18	361	588	36.7
19	283	567	93.4
20	121	552	86.1
21	105	552	86.0
22	70	541	80.1
23	105	552	6.0
24	300	536	90.3
25	1000	528	90.3
26	748	225	90.3
27	2400	165	813.8
28	1194	16.3	950.3
29	1194	14.7	580.7
30	1194	14.7	369.6
31	818	14.7	580.7
32	818	14.7	369.6
33	818	14.7	950.3
34	721	14.7	950.3
35	542	14.7	950.3
36	357	14.7	950.3
37	300	14.7	950.3
38	108	1.2	191.0

(1) All flow values are for one engine

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TABLE 4.2.1-1 (Cont'd)

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HEAT AND MASS BALANCE FOR GAS TURBINE COMBINED CYCLE POWER PLANT USING TEXACO OXYGEN-BLOWN GASIFICATION SYSTEM

39	108	30	191.0
40	108	30	4.8
41	108	115	4.8
42	108	115	4.8
43	108	30	186.2
44	178	30	186.2
45	178	30	172.8
46	178	30	13.4
47	250	30	13.4
48	250	30	186.2
49	255	2837	186.2
50	485	2837	186.2
51	485	2553	168.3
52	485	600	3.1
53	485	600	14.8
54	662	2400	168.3
55	662	2400	68.0
56	662	2400	100.4
57	662	2400	68.0
58	662	2400	100.4
59	662	2400	168.3
60	1000	2400	168.3
61	653	60 0	168.3
62	485	600	14.8
63	485	600	3.1
64	634	600	186.2
65	1000	540	186.2
66	1000	540	70.3
67	1000	540	115.9
68	108	1,2	115.9
69	108	1.2	70.3
70	108	1.2	186.2
71	485		3.1

					TAI	3LE 4.2.1-	.						1.1-25
		X	ATERIAL BA	LANCE FOR	TEXACO G	ASIFICATIC	N/ COMBINI	ED CYCLE 1	PONER PLA	NT			545-5
Stat	ton Number		1	2		e		4		ۍ . ا		9, 1	
Comp	. M.U.	Lb/Hr ⁽¹⁾	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Coal Lb/Hr	reed Mols/Hr	Lb/Hr Lb/Hr	Gas Mols/Hr
ਸ਼	16.04											321	20
H2	2.016											10632	5274
8	28.01											215201	7683
833	44.01											71208	1618
Η2S	34.08											6305	185
SS	60.07											541	6
02	32.00	713312	22291	599392	18731	138880	4 340	138880	4340				
N2	28.02	2321933	82867	1951117	69633	1681	60	1681	60			3699	132
<	39.94	41578	1041	34 9 48	875	661	20	661	20			66 <i>L</i>	20
H ₂ 0 Total	18.02	19534 3096357	<u>1084</u> 107283	19534 2604991	<u>1084</u> 90323	- 141360	- 4420	-	4420	168250		<u>58763</u> 367469	<u>3261</u> 18202
(1)	All values	based on	one engine										

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					TVT	1.2.4 270	-7 / (0/11						1-
			MATERIAL	BALANCE F	OR TEXACO	CASTFICA	TION/COMB.	INED CTC	LE POWER	PLANT			5554
Stati	on Number		2	æ	12	[13	1	14	T	S	16	۲-5
Comp.	M.U.	Lb/Hr ⁽¹⁾	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr
GH &	16.04	289	18	385	24	32	7	24	3	417	26	417	26
H2	2.016	9570	4747	13116	6506	1062	527	1062	527	14181	7034	14181	7034
8	28.01	193661	6914	265479	9478	21512	768	21512	768	286990	10246	286990	10246
₿ 2	44.01	64123	1457	87888	1997	2130	162	7130	162	94974	2158	94974	2158
S2H 4	34.08	5657	166	7770	228	613	18	613	18	8418	247	8418	247
88 . 2-7	60.07	481	¢	661	11	60	1	60	7	721	12	721	12
02	32.00	ł	1	I	I	ł	1	I	ŧ	I	I	ł	I
N2	28.02	3334	119	4567	163	364	13	364	13	4932	176	4932	176
×	39.94	719	18	666	25	80	2	80	2	1078	27	1078	27
H ₂ 0 Total	18.02	<u>52889</u> 330723	2935 16382	81991 462856	<u>4550</u> 22982	<u>13020</u> 43873	<u>326</u> 1819	<u>20957</u> 51802	<u>1163</u> 2656	<u>102948</u> 514659	<u>5713</u> 25639	<u>116301</u> 526012	6454 26380

TABLE 4.2.1-2 (Cont'd)

(1) All values based on one engine.

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TABLE 4.2.1-2 (Cont^{*}d)

MATERIAL BALANCE FOR TEXACO CASIFICATION/COMBINED CYCLE POWER PLANT

e	Mola/Br		16	8	276	170	m		J .		10	579
7	Lb/Br		32	1401	12147	5794	180		1514		180	21248
8	Hole/Br	19	5288	7633	1342	15	Q		78	20	20	16291
Ň	Lb/Br	305	10600	213800	59061	511	360		2186	199	360	309050
21	Mols/Hr	19	5274	7683	1618	185	6		132	20	8	14970
	Lb/Br	305	10632	215201	71208	6305	541		3699	662	541	309231
0	Mols/Hr	19	5274	7683	1618	185	6		132	20	48	14988
7	Lb/Hr	305	10632	215201	71.208	6305	541		3699	662	865	309555
•	Mols/Hr	19	5274	7683	1618	185	6		132	26	1501	16461
'n	Lb/Br	305	10632	21 5201	71208	6305	541		3699	799	27048	335738
80	Mols/Hr	Q	1760	2564	540	62	e		44	2	1615	6601
17-1	Lb/Hr ⁽¹⁾	96	3548	71818	23765	21:3	180		1233	280	29102	132135
Number	м. ч.	16.04	2.016	28.01	44.01	34 18	60.07	32.00	28.02	39.94	18.02	
Stat ion	Comp.	ង្វី	Н2	8	83	R2S	SOS	02	N S	<	и о И	Total

(1) All values based on one engine

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TABLE 4.2.1-2 (Cont'd)

MATERIAL BALANCE FOR TEXACO GASIFICATION/COMBINED CYCLE POWER PLANT

Station	Number	24-2	26	2	-	28-3 GT Exhau	37 ust Gas	38-7 Stei	
Comp.	м.и.	Lb/Hr ⁽¹⁾	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr 1	Mols/Hr
CH 4	16.04	305	19						
H ₂	2.016	10600	5258						
8	28.01	213800	7633						
822	44.01	59061	1342	395826	8994	395826	8994		
H2S	34.08	511	15	511	15	5.1	15		
SOS	60.07	360	9	360	9	360	Q		
02	32.00			391904	12247	505824	15807		
Ň	28.02	2186	78	1953302	69711	2324119	82945		
' 4	39.94	661	20	35706	844	42376	1061		
H ₂ 0	18.02	37247	2067	152215	8447	152215	8447	687661	38161
Total		345937	16438	2929824	100314	3421231	117275	687661	38161

(1) All values based on one engine.

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TABLE 4.2.1-3

PERFORMANCE SUMMARY FOR IGCC POWER PLANTS

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	Texaco	BGC
Gas Turbine		
Number of turbines	4	4
Air flow/turbine, lb/sec	860	860
Pressure Ratio	12	12
Combustor exit temp., F	2400	2400
Exhaust temp., F	1194	1177
Stack temp., F	300	335
Steam Turbine		
Throttle reheat temperature, F	1000	1000
Throttle pressure, psia	2400	2400
Condenser pressure, in. Hg	2.5	2.5
Overall System		
Gas turbine power, MW	630.4	605.6
Steam turbine power, MW	324.8	244,4
Net total power, MW	946.4(1)	819.2
Net heat rate, Btu/kWh ⁽²⁾	8716	8400
Coal rate, tons/day (as received)	8076	6752

(1) Includes fuel gas expander.

(2) Coal in-to-kW out at busbar.

FIG. 4.2.2-1



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TABLE 4.2.2-1

HEAT AND MASS BALANCE FOR GAS TURBINE COMBINED CYCLE POWER PLANT USING BGC GASIFICATION SYSTEM

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Station Number	<u>Temperature</u> (⁰ F)	Pressure (psia)	Flow (1b/sec) ⁽¹⁾
1	59	14.7	860.0
2	655	175	723.5
3			39.1
4	282	325	72.9
5	262	320	70.1
6	250	320	68.5
7	202	315	64.8
8	105	305	62.8
9	70	305	2.7
10	70	295	59.9
11	242	290	65.2
12	2400	167	788.7
13	1176	16.2	925.2
14	1176	16.2	601.9
15	1176	16.2	323.3
16	894	14.7	601.9
17	894	14.7	323.3
18	894	14.7	925.2
19	702	14.7	925.2
20	559	14.7	925.2
22	335	14.7	925.2
23	108	1.2	137.6
24	108	30	137.6
25	108	30	18.0
26	250	30	18.0
27	108	30	119.6
28	250	30	137.6
29	255	2837	137.6
30	485	2837	137.6
31	448	415	6.9
32	485	2837	130.8
33	448	415	6.9
34	662	2400	130.8
35	662	2400	130.8
36	1000	2400	130.8
37	657	600	130.8
38	657	600	130.8

(1) All flow values are for one engine.

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TABLE 4.2.2-1 (Cont'd)

HEAT AND MASS BALANCE FOR GAS TURBINE COMBINED CYCLE POWER PLANT USING BGC GASIFICATION SYSTEM

391000540130.840100054031.941100054098.9	<u>ec)</u> (1)
40 1000 540 31.9 41 1000 540 98.9	
41 1000 540 98.9	
42 935 415 4.8	
43 602 100 7.4	
44 108 1.2 86.7	
45 108 1.2 31.9	
46 108 1.2 118.5	

(1) All flow values are for one engine.

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TABLE 4.2.2-2

MATERIAL BALANCE FOR BGC GASIFICATION/COMBINED-CYCLE POWER PLANT

tat lor	n Number	-			0	ς,	_	4			Ś		ę
ċ	м. м.	Lb/Hr ⁽¹⁾	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr Coal	Mols/H r Feed	Lb/Hr	Mols/Hr	Lb/Hr	Mole/Hr	Lb/Hr	Mols/Hr
	16.04							15735	981	15735	981	15735	981
	2.016							6893	3419	6893	3419	6893	3419
	28.01						-	182793	6526	182793	6526	182793	6526
	44.01							9594	218	9594	218	9594	218
	34.08							5282	155	5282	155	5282	155
	60.07							420	7	420	2	420	2
	32.00	713312	22291	599392	18731			ł	ł	1	1	1	ł
	28.02	2321933	82867	1951117	69633			4259	152	4259	152	4259	152
	39.94	41578	1041	34948	875			80	7	80	2	80	3
_	18.02	19534	1084	19534	1084			37247	2067	26868	1491	21354	1185
l		3096357	107283	2604991	90323	140724		262303	13527	251924	12951	246410	12645

(1) All values based on one engine

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		LAM	FERIAL BAI	ANCE FOR	t BGC GAS	IFICAT	ION/COMB IN	NED-CYCLI	POWER	FIANT			
Stat ion	Number		7		ø	-	6	Ĩ	0	Т	ч		21
Comp.	М. М.	Lb/Rr ⁽¹⁾	Mols/Hr	Lb/Br	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Hr	Lb/Hr	Mols/Br
ਸ਼ੂ	16.04	15735	981	15735	981	176	11	15559	970	15559	970	8	1
в З	2.016	6893	3419	6893	3419	Ø	4	6885	3415	6885	3415	ł	ł
8	28.01	182793	6526	182793	6526	392	14	182401	6512	182401	6512	ł	ł
82	44.01	9594	218	9594	218	1320	30	8274	188	8274	188	337557	7670
H ₂ S	34.08	5282	155	5282	155	4873	143	409	12	604	12	ł	1
SOS	60.07	420	7	420	7	126	3	300	Ś	300	Ś	ł	ł
0 ₂	32.00	i	1	ł	I	ł	ł	ł	1	1	1	378464	11827
N N	28.02	4259	152	4259	152	2858	102	1401	50	1401	50	1952518	69683
¥	39.94	80	5	80	7	1	ł	80	2	80	7	34987	876
Н ₂ 0	18.02	8217	456	757	42	685	38	72	4	19119	1061	135150	7500
Total		233273	11916	225813	11502	10432	346	215381	11158	234428	12215	20 -38676	97556

(1) All values based on one engine.

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TABLE 4.2.2-2 (Cont⁴d)

4.2-15

MATERIAL BALANCE FOR BGC GASIFICATION/COMBINED-CYCLE POWER PLANT

23-47 Steam	Lb/Hr Mols/Hr										27504	27504	
t Gas	Mols/Hr				7670			15387	82917	1043	7500	114517	
13-22 GT Exhaus	Lb/Hr ⁽¹⁾				337557				492384	41657	135150	3330082	
Number	м. ч.	16.04	2.016	28.01	44.01	34.08	60.07	32.00	28.02	39.94	18.02		
Station	Comp.	CH,	H2	8	83	H ₂ S	cos	02	N2	¥	Н20	Total	

(1) All values based on one engine.

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4.2.3 Comparison of Performance

The performance estimates in Table 4.2.1-3 indicates that the IGCC power plant using the BGC gasifier would be 1.4 percentage points or approximately 3.5 percent more efficient than the Texaco-based IGCC power plant (40.6 versus 39.2 percent). The major contributor to this performance difference is the efficiency of the gasifier.

A review of Tables 4.2.1-1 and 4.2.2-1 show that the amounts of oxygen and steam used by the BGC gasifier are 26 and 50 percent lower than the oxygen and water used by the Texaco gasifier. The lower the oxygen and steam requirement, the more efficient the gasifer. A second benefit is that the oxygen plant and the boost compressor for the BGC-based system requires less power.

4.2.4 Off-Design Operation of IGCC Power Plants

The performance estimates presented thus far are for design point operation, i.e., full load at 59 F, 1000 ft (ISO) conditions. While the efficient power plants such as the IGCC power plants would be used as base-load plants (5000 h/yr or more) operating at full load, even these plants must follow load and be able to drop load, start, etc.

The IGCC power plant is a very complex plant having the characteristics of a chemical plant superimposed upon a power plant. The ability to successfully operate such a plant can only be demonstrated in an actual installation. It is possible, however, to simulate the part-load and transient behavior of this type of plant. Examples of this can be found in the literature (Refs. 4.2.-2 and -3) and have been the subject of a recent UTC in-house investigation (Ref. 4.2-4). The pertinent results of these studies are discussed below.

There are two modes of IGCC control: gasifier-lead in which coal/oxident flow to the gasifier is used to control load; and turbine-lead in which the turbine fuel controls are used to control load. Either control mode can be used successfully, i.e., either can meet the usual load ramp rates expected of large central station power systems. These load variations include the gradual changes for daily load following (4 percent/min from 80 percent of full load to full load and back) to rapid changes for tie line thermal backup (20-percent step change).

In transient and part-load operation, there are several critical operating parameters that must be observed, these involve the sulfur removal system and the heat exchangers. The Selexol absorber is a well proven process that operates unattended in some applications. However, in transient and part load operation, it is important to maintain the sulfur removal level so that emissions are within prescribed limits. To do this, the pressure control point for the overall systems should be directly downstream of the Selexol unit. This will allow close control of the pressure in the absorber during all operations.

As load is changed, the turbine inlet temperature and the exhaust temperature also change. The fuel gas temperature from the gasifier, however, remains essentially constant. Steam temperature from the waste heat boiler located in the fuel gas stream remains constant but the pinch temperature (minimum approach) between the exhaust gas and the steam in the superheater and reheater changes. It is necessry to balance the flows so that an adequate temperature difference is maintained, i.e., the steam is always cooler than the exhaust gases. The pinch temperature at the HP and LP boilers (see Fig. 4.2.1-1) are shown in Table 4.2.4-1.

Of particular interest is the performance at part load. Combined-cycle systems have excellent part-load efficiencies and the IGCC systems continue to display this characteristic. The part-load heat rate for the IGCC plant is shown in Fig. 4.2.4-1. Here it can be seen that at 70-percent load, the heat rate has increased by only 8.9 percent. Also shown in Fig. 4.2.4-1 is the relationship between steam turbine output and load. As indicated in Table 4.2.4-1, reduction in load reduces throttle conditions and, thus power. L



PART-LOAD PERFORMANCE OF IGCC POWERPLANTS

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TABLE 4.2.4-1

PINCH TEMPERATURE IN TEXACO IGCC POWER PLANT

	Pinch 1	Cemp., F	Throttle Conditions			
Load, Percent	LP	HP	Temp.,F	Press., psia		
100	40	40	1000	2400		
90	45	31	935	2157		
80	50	22	864	1911		
70	54	14	785	1664		

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4.2.5 References

- 4.2-1 Interim Project Briefing, NASA-MSFC, Huntsville, Alabama, December 17/18 1980.
- 4.2-2 Carlson, N. G.: High Temperature Turbine Technology Program, UTC Final Report FE-2292-19, April 1977.
- 4.2-3 Fluor Engineers and Constructors: Entrained Gasification Combined-Cycle Control Study - Vol. I. EPRI AP 1422, June 1980.
- 4.2-4 Lehman, S. J.: Off-Design Performance Variation of an Advanced Combined Cycle Power Plant Using the V84.3 Gas Turbine and the Texaco Entrained Flow Gasifier. UTRC Report R80-752717-1, December 22, 1980.

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1.3 Assessment of MHD Technology Status

Magnetohydrodynamics (MHD) is a branch of physics which deals with phenomena arising from the motion of electrically conducting fluids in the presence of electric and magnetic fields. This covers a broad area ranging from astrophysics to a number of technological applications including power generation, propulsion, pumping, flow measurement and magnetic - confinement thermonuclear fusion. The types of conducting fluids include liquid metals, electrolytic solutions and ionized gases or "plasmas."

This assessment is restricted to the application of MHD to the production of electric power by utilizing an ionized gas as the conducting fluid. There are basically two categories of cycles for MHD power generation; open-cycle and closed-cycle. In open-cycle MHD power generation, the ionized gas flows through an MHD generator channel and is ultimately exhausted to the atmosphere after treatment to make it environmentally acceptable. In closed-cycle MHD power generation, the ionized gas is continuously recirculated through the MHD channel in much the same manner as the gas in a closed-cycle gas-turbine system. The assessment is further restricted to consideration of open-cycle MHD power generation because it is in a more advanced state of development and appears to be much closer to commercial application. Also, the majority of the financial support for MHD power generation programs in the U.S., provided by the U.S. Department of Energy and the Electric Power Research Institute, are primarily directed towards open-cycle MHD power generation.

1.3.1 Assessment of MHD Technology Readiness Development Trenus

The basic principles of the phenomena involved in MHD power generation have been known since experiments in electrodynamics were performed by Michael Faraday in the nineteenth century. However, the only electrically conducting fluids which were practical at that time were liquid metals and electrolytes. Developments in this century in high temperature technologies led to interest in gases as conducting fluids. Experiments with gaseous MHD power generation were conducted between 1938 and 1944 by Westinghouse, with unsatisfactory results. However, as the technologies associated with high temperature fluid dynamics advanced, interest was again arrused in MHD power generation. Experiments were conducted in several industrial research laboratories in the late 1950's and early 1960's. These experiments have yet to lead to commercially available equipment. A review of selected efforts, however, will establish the current technology base.

1.3.1.1 Technology Readiness - Earlier Programs

In 1959 a test was conducted at Avco-Everett Research Laboratory in which 11.5 kW of electric power was generated for a period of 10 seconds. In 1960, a 10-kW MHD generator was operated at Westinghouse for a period of over one hour. General Electric Co. was also active in MHD during this period. These early tests produced only small amounts of power, but the demonstrated the possibility of

generating power by MHD and considerable interest was displayed by industry, government, and universities, both in this country and in several other countries. 1

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In 1965, Avco-Everett Research Laboratory performed a test in which 32 MW of electric power was produced in an MKD generator for a period of one minute. The duration was limited by the capabilities of the test facility rather than by failure of the generator. Another generator built by Avco produced 18 MW for one minute in 1966. Long duration testing at Avco was conducted on a much smaller scale. One Avco long duration test facility had accumulated over 2000 hours of operation by 1966 at a power level of several kW.

Another development in the 1960's which has had a significant impact on the potential for MHD power generation is the progress made in constructing superconducting magnets. This field was in its infancy in 1961. Considerable progress in superconducting magnet development has been made because of its applicability to high energy physics experiments, controlled thermonuclear fusion and other areas as well as MHD power generation.

MHD programs have been initiated in several other countries. France, The Federal Republic of Germany and Great Britain developed strong programs, but they eventually declined as these countries shifted emphasis to nuclear energy. However, MHD programs have continued to grow in many other countries, including the U.S.S.R., Japan, China, the Netherlands, Poland, India, Australia and Italy. To, v, the U.S.S.R. has the largest MHD program in the world. In 1965, they corrected construction of their U-02 facility which has a thermal power input apacity of 5 MW. In 1971 they completed construction of their U-25 facility which has a thermal power input capacity of 300MW.

In the decade of the 1970's, the MHD program in the United States has grown considerably, with a shift in emphasis towards coal as a fuel. Funding of MHD by the Department of the Interior's Office of Coal Research was at a level of several million dollars per year in the early 1970's. Funding has steadily increased as the MHD program came under the jurisdiction of the Energy Research and Development Administration and, later, under the Department of Energy. The DOE appropriations for the MHD program were 79 million dollars in fiscal year 1980 and 67 million dollars for FY 81.

Great strides in MHD have been made in the past decade. An MHD channel has been operated at a power level of several hundred kWe for two continuous 250-hour runs Although this channel is small compared to that required for a commercial power plant, the local conditions at the electrodes of the channel were comparable to those anticipated for a commercial-scale channel. In the U.S.S.R., a channel has been operated for 250 hours at a power level of several MWe. In other tests, electric power was successfully delivered to the Moscow electrical network on a small scale. No MHD power plants have been constructed to date which produce electric power specifically for utility use. The earliest date by which such a plant could be built in the U.S.S.R. is probably 1985. The earliest date by which such a plant could be built in the U.S. is probably 1990. The construction of an MHD power plant in the U.S. could probably be substantially accelerated by retrofitting an existing steam power plant by the addition of an MHD generator as a topping unit. The actual dates of implementation of MHD for the production of electric power depend, of course, upon further technological advancement and upon the support given to research and development by government and industry.

1.3.1.2 Technology Readiness - Projected Commercial Plant

A general summary of the major characteristics of a projected commercial MHD power plant is indicated in Table 1.3.1.2-1. Since no such power plants have been constructed, these characteristics are based upon conceptual designs, anticipated technological developments and extrapolation of test data. The power plant would be a combined MHD/steam power plant with a net output in the range of 500 to 2000 MW-electrical. Approximately one half of this power would be produced by the MHD generator and the remainder by the steam-turbine driven generator. The large size of the plant is desired because of the tendency for the efficiency of an MHD generator to increase significantly with size. The increase in efficiency as size (and hence the volume to surface-area) increases is due to the fact that the power generating phenomenon is a volumetric effect while the thermal, viscous and electrical losses are surface effects. The overal power plant efficiency from the coal pile to the utility grid is projected to be in the vicinity of 50%. This is the major reason for the interest being expressed in MHD power generation at the present time.

Emission levels from the plant are projected to be below the environmental limits. The collection of flyash and K_2SO_4 result in adequate control of particulate matter and sulfur-oxide emissions. Nitrogen oxides and carbon monoxide emissions can also be controlled in an MHD power plant by control of combustion conditions and the MHD generator exhaust gas cooling rate. Due to the high power plant efficiency, the heat rejection rate is less than that of conventional plants which have overall efficiencies of 40% or less.

Another attractive feature of an MHD power plant is the capability it has for direct utilization of coal as a fuel. Coal can be injected directly into the combustor and the resulting combustion products can be tolerated by the MHD generator. Problems which are typical of conventional coal fired plants do arise, but the presence of coal ash is not highly detrimental to the MHD generator itself.

Table 1.3.1.2-2 presents a summary of projected MHD/steam power plant performance results and cost estimates based upon several recent investigations. The investigations are the Energy Conversion Alternatives Study (ECAS) completed in 1975 (Kef. 1.3-1) and the Parametric Study of Potential Early Commercial MHD

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TABLE 1.3.1.2-1

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PROJECTED CHARACTERISTICS OF A COMMERCIAL-SCALE MHD/STEAM POWER PLANT

Size Range

500 - 2000 MWe

Ef 'cincy

50%

Environmental Control

Low SO_x , NO_x , CO, Particulate and Thermal Discharges

Fuel Capability

Coal (direct-fired)

Cost of Electricity

Competitive

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TABLE 1.3.1.2-2

SUMMARY OF PROJECTED MHD/STEAM POWER PLANT PERFORMANCE AND COST ESTIMATES

Source of Data	ECAS	PSPEC
Type of Plant	Direct-Fired Air Preheater	Oxygen-Enriched Air
Net Electrical Power Output (a.c.)	1932 MWe	931 MWe
MHD Generator Electrical Power Output (a.c.)	1406 MWe	512 MWe
Overall Efficiency	48.3 %	43.4 %
Year of Cost Basis	mid-1975	mid-1978
Plant Capital Cost	\$718 per kW	\$646 per kW
Cost of Electricity	31.8 mills/kWhr	32.5 mills/kWhr

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Power Plants (PSPEC) completed in 1979 (Ref. 1.3-2). The cost estimates shown for these studies are given in 1975 and 1978 dollars, respectively. The basis for the ECAS design is an MHD power plant utilizing advanced or "second generation" technology projected to be available for commercial orders in the mid to late 1990's. The basis for the PSPEC design is an MHD power plant utilizing near term or "first generation" technology projected to be available for commercial orders in the early 1990's.

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1.3.1.3 Trends in Operating Parameters

Figure 1.3.1.3-1 shows a simplified sketch of an MHD generator. The generator consists of a duct with two electrode walls and two insulating walls with a magnetic field applied perpendicular to the duct axis. When an ionized gas flows through the duct, electrical energy is extracted directly from the kinetic and thermal energy of the gas due to its interaction with the magnetic field. The current in the gas is carried almost entirely by electrons. The net current generated may pass through a single load as indicated in Fig. 1.3.1.3-1 or through a set of multiple loads, depending on the type of electrode interconnections.

The MHD generator essentially behaves as an electromagnetic "turbine", being thermodynamically equivalent to a gas turbine. Both types of machines can be most efficiently utilized for production of electric power when operating in a Brayton topping cycle over a conventional Rankine-cycle steam power plant. MHD generators, however, operate at much higher temperatures (2300 to 2800° K or 3700 to 4600° F), lower pressures (6 to 10 atm) and are considered to be most efficient at very large sizes (500 MWe).

Plasma Properties

The plasma properties of primary interest in MHD power generation are the electrical conductivity σ and the Hall parameter ωr of the gas. The plasma is essentially electrically neutral and the conduction of electricity in the plasma is carried out primarily by electrons. The Hall parameter relates to the effect of the magnetic field on the electrical conduction process.

In order to conduct electricity, the gas must be partially ionized, meaning that some of the atoms in the gas are stripped of one or more of their electrons, as occurs at high temperatures. If an alkali metal is added to the gas, sufficient ionization may occur at a reasonable temperature because of the metal's low ionization potential. Table 1.3.1.3-1 indicates the ionization potential of several materials. The process of adding an alkali metal to a gas to achieve the desired degree of ionization is referred to as "seeding". The primary candidate seed materials for use in MHD power generation are potassium (K) and cesium (Cs).

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FIG. 1.3.1.3-1



GENERATION OF D.C. ELECTRIC POWER BY MAGNETOHYDRODYNAMICS

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TABLE 1.3.1.3-1

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IONIZATION POTENTIAL OF METAL VAPORS

Metal	Ionization Potential (electron-volts)
Lithium	5.363
Sodium	5.12
Aluminum	5.96
Potassium	4.318
Calcium	6.09
Rubidium	4.16
Cesium	3.87
Barium	5.19
Mercury	10.39

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The electrical conductivity of the plasma depends upon the composition, temperature and pressure of the gas. Figure 1.3.1.3-2 shows the variation of electrical conductivity with temperature and pressure for a specific seed concentration and gas composition. For economical power production in an MHD generator, conductivity requires very high combustion temperatures. In combustion gases, the presence of the OH radical introduces the posibility of electron attachment, forming OH⁻, and reduction of electrical conductivity. Therefore, fuels with high carbon content are preferred over fuels with high hydrogen content. In general, however, it is difficult to predict plasma conductivities with a great degree of precision.

The Hall effect arises because of the tendency for charged particles to take on circular or spiral trajectories in a magnetic field. The cyclotron frequency ω of an electron of charge e and mass m_e in a magnetic field of strength B is given by:

$$\omega = \frac{eB}{m_e}$$

Collisions between free electrons and other molecules and atoms in the gas prevent the electrons from continuing to spiral indefinitely and cause the electrons to migrate in a direction dependent upon the relative strengths and directions of the electric and magnetic fields present. The Hall parameter is a dimensionless parameter equal to the product of the electron cyclotron frequency and mean time t between collisions of electrons with other particles:

Hall Parameter = $\omega \tau$

The Hall parameter is proportional to the magnetic field strength and is dependent upon the gas composition, pressure and temperature.

MHD Channel Phenomena

In the following paragraphs, the basic relations are developed for a simplified model of an MHD generator. This is a "zero-dimensional" model in which the variables (such as velocity, conductivity, electric field strengths, etc.) do not vary with respect to location within the generator. Although this is unrealistic, the model is useful for illustrating basic problems.

Figure 1.3.1.3-3 is an elementary diagram showing the electrode walls of an MHD duct (channel) with a connected external electrical load (shown simply as a resistor). The magnetic field is perpendicular to the diagram, directed away from the reader. The motion of the ionized gas induces an electric field across the duct in the direction from the anode to the cathode. The magnitude of this induced electric field is equal to the product of the velocity, u, and the magnetic field strength, B.

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MHD DUCT WITH CONTINUOUS ELECTRODES WITHOUT HALL EFFECT

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Einduced = uB

If the velocity has units of meters per second and the magnetic field strength has units of tesla (1 tesla - 10,000 gauss), the electric field will have units of volts per meter.

This induced field tends to make the cathode positive in voltage with respect to the anode. Thus if an external load is connected between the terminals on the electrodes, an electric current would pass through this load in the direction from the cathode to the anode. A corresponding electric current would pass through the gas in the duct in the direction from the anode to the cathode. This is shown in Fig. 1.3.1.3-3, where the current inside the duct is indicated as a "current density", j, which has the units of current per unit area (amp per square meter).

The magnitude of the voltage difference between the electrodes depends upon the magnitude of the external electrical load connected to the terminals as well as upon the induced field, uB. The occurrence of this voltage difference induces an electrostatic field in the gas opposite in direction to the magnetically induced field uB. Thus the net electric field is:

> E = E_{induced} = E* = uB - E*

where E* is the "electrostatic" field arising from the potential difference between the electrodes. The directions of these field components are shown in Fig. 1.3.1.3-3.

The electric current density, j, will be proportional to the electrical conductivity, σ , of the gas, which is expressed in units of mhos per meters (where a mho is the reciprocal of an ohm). If there is no "Hall effect", the electric current will be parallel to the net electric field and is given by the expression:

 $j = \sigma (uB - E^*)$

The electric power P generated per unit volume is equal to the product of the current density and the electric field through E*. Thus,

 $P = jE^* = \sigma E^* (uB - E^*)$

The Hall effect results from the combined tendency for electrons to spiral around the lines of magnetic flux and to interact with other particles. Thus, when an ionized gas is the presence of both a magnetic field and an electric field, the electric current is not parallel to the electric field. As shown in Fig. 1.3.1.3-4, there is an angle, Θ , between the electric field direction and the current direction. In terms of the dimensionless "Hall parameter", ω_T , the "Hall angle" is

 $\theta = tan(\omega\tau)$

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FIG. 1.3.1.3-4



MHD DUCT WITH CONTINUOUS ELECTRODES WITH HALL EFFECT

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where $\omega \tau$ is directly proportional to the magnetic field strength as previously indicated. Typically, 0 would be in the range of 45 to 75 degrees.

If the electrodes are continuous (that is, if they are continuous walls of constant electrical potential), the electric field and electric current directions are as shown in Fig. 1.3.1.3-4. The fact that the electric current traverses diagonally across the MHD duct results in a reduction in the power output of the generator. The magnitude of the reduction in power compared to the power which would be generated with no Hall effect is equal to the factor

 $1 + (\omega \tau)^2$

In order to counteract this detrimental effect, the channel can be constructed by "segmenting" the electrode walls, as shown in Fig. 1.3.1.3-5. To make the current traverse the duct in the direction perpendicular to the flow direction, each opposing pair of electrodes is connected to an individual external load. For this case, the electric field is directed diagonally across the duct, and the power output is equivalent to that which would occur with no Hall effect. This configuration is referred to as a segmented "Faraday" generator. Ì

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The disadvantage of this arrangement is that there must be a large number of individual loads. A commercial-scale duct would require approximately 1000 segments. However, an alternate method for making connections between electrode pairs which can reduce the number of loads is illustrated in Fig. 1.3.1.3-6. This is referred to as a "diagonal generator." Each electrode segment on the anode wall is connected to a corresponding electrode segment on the cathode wall through a short circuit. The load is then connected between the farthest upstream and farthest downstream electrode pairs. If the diagonal lines between interconnected electrodes (shown as dashed lines in Fig. 1.3.1.3-6) correspond to lines of constant electrostatic potential for the segmented Faraday configuration of Fig. 1.3.1.3-5, then the directions of the electric field and electric current are the same as that for the ideal case. The advantage of this configuration is that there is only one load. But once the interconnections have been made, the ideal power extraction can only occur at the corresponding value of the Hall parameter, wt. If the Hall parameter should happen to deviate from the design value, the generator performance will diminish.

The equations presented in this section were developed on the basis of a "zero-dimensional" model of the MHD channel. In reality, even though the magnetic field strength, velocity, conductivity and other parameters vary throughout the channel, the equations are applicable locally at any point within the channel. Figure 1.3.1.3-7 from Kef. 1.3-3 shows how the values of the parameters (averaged over the channel cross section) vary along the direction of the axis of the channel.

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FIG. 1.3.1.3-5



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SEGMENTED FARADAY GENERATOR

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DIAGONAL GENERATOR

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Channel Construction and Materials

There are a number of alternate means of constructing the MHD channel in order to meet design requirements. The channel construction approach which has been proposed by Avco-Everett Research Laboratory for the MHD Engineering Test Facility (Ref. 1.3-4) has been selected. Both the Faraday and Diagonal modes of in crelectrode interconnection will be considered.

Three variations of the Faraday NHD channel concept are shown schematically in Figs. 1.3.1.3-8(a), 8(b) and 8(c) from Ref. 1.3-4. The differences between the three are in the design of the insulator wall. The insulating walls have the requirement of preventing electric current from flowing along the walls in a direction perpendicular to the direction of the electrostatic field, E*. There are three ways to achieve this requirement: a continuous insulating wall, a peg-wall configuration, or a diagonal-bar configuration. The continuous insulating wall is simple in principle but puts a severe restriction on the choice of material of construction. The material must be a good electrical insulator and, of course, must be able to maintain its structural integrity and its insulating property during operation.

Figure 1.3.1.3-8 illustrates the multiple-load Faraday design using metallic peg wall construction. The wall is constructed of metal "pegs" which are insulated from one another, thus minimizing the flow electric current along the wall. However, heat can be conducted away from the surface of the wall to cooling water. In general, the segmentation of the pegs is about equal to that of the electrodes. Square pegs are illustrated but polygon cross sections can be and have been used. An allceramic side wall and peg wall are approximate electrical equivalents, but the feasibility of ceramic walls for long duration has not yet been demonstrated. In contrast, several promising peg wall schemes have been evolved and tested. 1

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The advantage of the peg-wall Faraday channel is that its performance is generally considered to be the best of all possible channel types. Its disadvantage is associated with the mechanical complexity and the question of long term reliability of a peg wall. The number of pegs is large. Provision for adequate and reliable cooling can be a formidable design problem, given the mechanical and electrical stress conditions present in an MHD generator.

Figure 1.3.1.3-8 shows another Faraday type channel in which the insulator wall is constructed of long bars rather than pegs. Some segmentation of the bars may be incorporated to control gap voltages and current densities. The bars are oriented along the equi-potentials predicted for the free-stream plasma. The gross performance of the Fig. 1.3.1.3-8(b) channel is equivalent to that of the peg wall type of Fig. 1.3.1.3-8(a), at the full load or design point. The advantage of a bar wall is the mechanical simplicity and reliability of this kind of construction. Its disadvantage is in part-load (or off-design) operation where the fixed bars will be at an incorrect angle. The resulting circulation of Hall current is likely to make performance less than that of a part-load, peg wall design.

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FIG. 1.3.1.3-8



MHD CHANNELS WITH FARADAY INTERCONNECTION AND ALTERNATE INSULATING WALL STRUCTRURES

A possible compromise for side wall construction is depicted in Fig. 1.3.1.3-8(c). In this concept the bars are highly segmented but the segmentation is not as great as that of a peg wall. The approach here is to approximate a peg wall design with its good part-load performance, and yet retain some of the mechanical simplicity of bar wall construction. This concept needs further investigation within the context of part-load performance requirements.

Variations of the basic single-load diagonal channel are illustrated in Figs. 1.3.1.3-9(a) and 9(b) from Ref. 1.3-4. Figure 1.3.1.3-9(a) shows the "window frame" design in which each electrode/bar pair is an integral unit; no external connections are necessary. The window frame technique appears to represent a simple and low cost type of construction. However, window frames may have a serious fault in that severe current concentrations tend to exist on the bars near the corners. These current concentrations cause unacceptable long term damage as has been observed in experiments. To alleviate this difficulty, the current path is interrupted by separating the bars from the electrodes, and each bar is split into at least two elements. The electrode pair previously connected by virtue of the window frame technique must now be connected by external wires. This arrangement is shown in Fig. 1.3.1-9(b). In this design the window frame technique cannot be used, but the basic mechanical simplicity of a diagonal channel is retained.

There are two disadvantages to the single load diagonal channel concept. First, ideal single load diagonal channel performance will be less than that of the multiloaded Faraday machines because the loading cannot be adjusted to an arbitrary distribution along the channel's length. Secondly, part-load performance may be significantly less than that of any of the Faraday options. Multiple loading offsets these disadvantages. Two multiple loading schemes are depicted on Figs. 1.3.1.3-10(a) and 10(b) from Ref. 1.3-4. These apply to either window frames or split bar designs.

The design of the electrodes and insulators and the selection of materials of construction are determined by the requirements for efficiency and material durability and the severity of the environment to which they are exposed and the electrical phenomena occurring in the immediate vicinity of the walls. The high temperature of the plasma dictates that a means be provided to cool the wall elements, but not to such an extent that large heat transfer losses occur. From the viewpoint of phenomena occurring in the ionized gas itself, it is desirable to have the gas-side surfaces of the wall elements at high temperature. The wall elements must be able to withstand corrosion and erosion by oxidation and by electrolytic action as well as damage by arcing phenomena. Consideration must also be given to efficient transfer of electric current from the plasma to the external load through the elements and compatibility of materials with one another and with their environment.



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DIAGONAL MHD CHANNELS WITH MULTIPLE LOADS



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An approach which has been established in the U.S. is to employ the principle of channel wall protection by deposition of a maintenance material on the interior surfaces of the channel. In the early 1960's, MHD channels were built and tested in which zirconia was injected into the gas stream to form a protective coating. The principle was to establish an equilibrium state in which the rate of erosion of this coating would equal the rate of deposition. More recently, coal slag has been utilized as the maintenance material since it is naturally present in the products of combustion in a direct coal-fired NHD plant. Experiments with coal-slag deposition have indicated that this is a feasible approach and is not detrimental to MHD generator performance or operation. Figure 1.3.1.3-11 schematically shows a channel in which the electrode walls are covered by a protective coating of a material which is continually eroded and replenished. Figure 1.3.1.3-12 schematically shows the various regions of the channel including the plasma and the electrode wall elements. Figure 1.3.13-13 from Ref. 1.3-5 illustrates a number of electrode system design configurations showing the different types of materials required in each design in order to fulfill the numerous requirements of the electrodes.

A specific electrode design, which has evolved at Avco after extensive testing over a period of several years with deposition of maintenance materials simulating the properties of coal slag, is illustrated in Figs. 1.3.1.3-14 for an insulating peg-wall design and 1.3.1.3-15 for an insulating bar-wall design. The basic electrode design is a water-cooled copper bar, clad with a thin layer of oxidation resistant material over vulnerable surfaces. Cladding materials under consideration include platinum, paladium, gold, silver, inconel and nickel. Extensive long duration testing is necessary to determine which materials are best, considering both technical performance and cost. The plasma-side surface of the electrode has grooves, prefilled with ceramic (e.g., zirconia), to provide attachment points for the deposited slag and thereby facilitate development of a slag layer. The high thermal conductivity and high thermal diffusivity of the copper substrate makes it more suitable as an electrode than other materials. Specifically, the local high heating associated with arcing phenomena is more readily dissipated in copper. Arc damage is thereby alleviated or avoided. Boron nitride (BN) hard spacers are employed as insulators since BN is a good thermal conductor, is resistant to thermal shock, and is not wetted by coal slag.

The design for insulator walls is the same as that for the electrodes, and essentially for the same reasons. Namely, water cooled copper peg or bar-segments are used, with grooves on the gas side surface to facilitate slag layer development. Dense aluminum oxide spacers may be used as gap insulators rather than BN to minimize cost. The various erosion difficulties are not as severe on insulator walls compared to electrode walls. But in the special case of a peg wall, experimental evidence indicates that BN is necessary. BN's capability to be better cooled than Al_2O_3 , and its good resistance to thermal shock affords better protection of the numerous O-ring seals associated with a peg wall.

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(b) COAL-FIRED SYSTEM

MHD CHANNELS WITH AND WITHOUT MAINTENANCE LAYERS

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FIG. 1.3.1.3-12

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ELECTRODE SYSTEM CINFIGURATIONS

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FIG. 1.3.1.3.-14



ELECTRODE AND PEG WALL DESIGN

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ELECTRODE AND BAR WALL DESIGN

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1.3.2 Develop Data Base For MHD Programs

Because MHD systems are far from commercialization, there is no operational data base. It is possible, however, to review the development programs currently underway. An MHD power plant will include a number of components which require further research and development before they can perform satisfactorily. These include the MHD channel, superconducting magnet, inverter system, combustor, air preheater system, heat and seed recovery system and the seed regeneration system. In general, each of these components and systems must be scaled up to larger sizes than those which have been tested to date. Their performances in general require improvement and longer lifetimes must be demonstrated. In most cases, the performance is expected to improve as the components are scaled up. Further advances in system integration are also required as very few facilities presently exist in which all of the major components have been tested together as a unified system.

1.3.2.1 Major Test Facilities

A number of facilities exist in the U.S. for testing of specific MHD components at various scales. Two major facilities are now under construction for larger scale testing and for integration of components. These are the Coal Fired Flow Facility (CFFF) at the University of Tennessee Space Institute and the Component Development and Integration Facility (CDIF) in Butte, Montana. The objective of both facilities is to provide test data for long-duration operation at significant scales. Another major facility is the High Performance Demonstration Experiment (HPDE) facility at the Arnold Engineering and Development Center (AEDC).

The CFFF is a new facility which will initially have the capacity for a 20-MW thermal input and later will be increased to a 50-MW thermal level. Three test bays will be installed which will give the facility the capability for testing various combustor, channel and magnet configurations. Downstream equipment will be installed for testing steam bottoming plant components and recovery of seed from the gas stream. The CDIF will also be a coal-fired facility and will have a 50-MW thermal input capacity. Emphasis will be upon testing of MHD flow train components under conditions of low slag carryover (approximately 10 percent) into the channel. Both facilities are scheduled for initial startup in 1981.

The HPDE at AEDC is designed for short-duration operation (10 to 20 seconds) and began operation in 1979. Tests were performed in 1979 and 1980. The channel is now undergoing modification and will be started up again in 1981.

1.3.2.2 MHD Channel Technology

The principal design problems associated with the MHD channel are power extraction and durability. The electric power must be extracted from a multitude of electrodes and it must be controlled, conditioned and inverted to alternating current. The walls of the channel must resist a unique combination of electrochemical, electrodynamic, thermal and aerodynamic stresses. Effective power

extraction from MHD channels has been demonstrated at power levels up to 32 MWe in the U.S. and up to 20 MWe in the U.S.S.R. The results have given researchers confidence that large sized channels can perform according to expectations.

Long duration testing has been conducted at lower, yet significant, power levels. A U.S. channel has been operated by Avco for two continuous 250-hour test runs with a power output of several hundred kilowatts. Ash was injected into the combustor to simulate coal firing. Local electrical conditions near the electrode were comparable to conditions which would be anticipated in a larger scale channel. Testing of a channel with direct coal firing in the combustor has been achieved at the University of Tennessee Space Institute. In the U.S.S.R. a channel has been operated for 250 hours with the power output ranging from 3 to 11 MW_o.

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The major organizations involved in MHD channel design, construction, testing or analysis include Avco-Everett Research Laboratory, University of Tennessee Space Institute (UTSI), Modern Electric Power Products and Services Company (MEPPSCO), Reynolds Metals Company (RMC) and STD Research Corporation. Avco has built and tested numerous generators including the largest tested to date. They have recently delivered a channel to CDIF site, along with a combustor of Avco's design. UTSI has built and extensively tested small channels utilizing coal as the fuel. Upon completion of the CFFF, UTSI capabilities will be greatly expanded. RMC is presently operating an MHD generator test facility. MEPPSCO has designed several very large generators, including the channel for the HPDE and a channel for the U-25 facility in the U.S.S.R. STD has developed extensive computer programs for analysis of MHD channel phenomena and behavior.

1.3.2.3 Superconducting Magnet

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Superconducting magnet technology has experienced advances because of the wide application of large superconducting magnets. A magnet for an MHD plant requires a nearly uniform magnetic field of approximately 6 Tesla across an elongated channel. This is most effectively achieved by means of a saddle-shaped coil. The largest MHD superconducting magnet constructed to date is one which was manufactured in the U.S. and tested in the U.S.S.R. The magnet system weighed approximately 40 tons, produced a field of 5 tesla and had a stored energy capacity of 34 megajoules. The magnet has been operated successfully and reliably.

For commercial application in a 1000-MW_e power plant, the magnet would have to be scaled up to approximately a 5000 megajoule stored energy capacity. The major difficulties anticipated in scaling up to this level is in the area of structural integrity in the presence of very large magnetic forces, construction techniques for the coils and keeping the magnet cost within reason. This is viewed as a design problem rather than a technology problem as no technological breakthroughs are required.

The DOR MHD superconducting magnet programs are being conducted under management of the Francis Bitter National Magnet Laboratory at MIT. Organisations which have built or are building magnets for MHD facilities include Argonne National Laboratories, Magnetic Corporation of America and G.E.

1.3.2.4 Inverter Systems

Inverter technology is another area in which advances have been made without assistance from the MHD program. Application of inverters in MHD systems has been very limited to date. In most of the experiments, the generated electrical power is dissipated in a resistive load. In the U-25 facility in the U.S.S.R., MHD power has been fed to the Moscow grid by means of inverters. The cumulative time over which this has occurred to date is more than 800 hours.

Inverter equipment is presently available. The major area in which development is required is in designing an inverter system to meet the requirements of an MHD channel. The Electric Power Research Institute (EPRI) is playing a major role in the inverter program for MHD. EPRI is responsible for the procurement of the inverter system which will be installed in the Component Development and Integration Facility in Butte, Montana. Avco-Everett Research Laboratory, Westinghouse, General Electric Company and the University of Tennessee Space Institute are engaged in work on MHD inverter systems.

1.3.2.5 Combustors

MHD combustors must provide a very high temperature gas with a rather uniform flow with uniform mixing of seed to the MHD channel. Combustor operating pressures must be in the range of five to ten atmospheres. A major criterion for combustor performance is minimum heat loss through the walls of the combustor. When coal is used as a fuel, the coal slag must be handled, as well. One major coal-fired combustor approach is to remove up to 90 percent of the slag from the combustor. Another major approach is to allow all of the slag to be carried over into the MHD channel.

A combustor utilizing clean fuel with a thermal power input of 250 MW has been successfully operated in the U.S.S.R. Combustors utilizing coal as the fuel have been operated in the U.S. inlcuding an $8-MW_t$ combustor without slag removal and a $5-MW_t$ combustor which achieved 95-percent slag removal. Future development programs are aimed at scaling up present-day combustor designs and reducing heat losses.

The Pittsburgh Energy Technology Center is responsible for the DOE MHD combustor programs. Avco, TRW and Rockedyne Division of Rockwell International have all developed combustor designs and have built and tested them at the 20-MW, level.

1.3.2.6 Air Preheater Systems

Regenerative air preheater systems of the type required for MHD power plants have been utilized in the steel industry for many years. Their temperature of operation, however, is below that required for MHD systems. Regenerative air preheaters for MHD have been constructed and tested. Preheaters operating on clean fuels have been tested in the U.S. for over 100 hours with air outlet temperatures over $3000^{\circ}F$. In the U.S.S.R., separately fired air preheaters have operated as part of an MHD pilot plant, with air preheat temperatures up to $2240^{\circ}F$.

Scaling up separately-fired high temperature air heaters to commercial power plant size appears to be achievable provided clean fuels are utilized. The challenge is in the area of developing a coal combustion system which can provide sufficiently clean combustion products to fire the heaters. Direct fired air preheaters represent a greater benefit in terms of plant efficiency as well as a greater challenge in terms of development. Materials capable of withstanding the severe seed/slag environment of a direct-fired air preheater system are currently being investigated. Extensive testing programs and materials analysis programs are in progress. Air preheating elements have undergone over 800 hours of testing in a slag/seed environment at temperatures up to 2780°F. Substantial programs on high temperature air preheater development have been undertaken by FluiDyne Engineering Corporation, Avco and G.E.

1.3.2.7 Heat Recovery/Seed Recovery Systems

The heat recovery system includes the heat transfer surfaces which transfer heat from the hot gas flow stream to the steam bottoming plant boiler feedwater and produce steam for driving the steam turbines. The seed recovery system includes the surfaces and equipment which contribute to the removal of seed compounds from the hot gas flow stream. Basically, these systems are comprised of conventional steam plant boiler equipment and particulate removal systems such as electrostatic precipitators. However, the seed-laden environment presents a considerable challenge in terms of prevention of corrosion, erosion and plugging as well as designing the system to optimize the collection of seed.

Testing programs to simulate MHD plant gas compositon an flow conditions in heat recovery equipment are being conducted at Mississippi State University and Argonne National Laboratory. The Department of Energy has issued a contract to Babcock and Wilcox Company for the design, construction and installation of a larger heat-recovery/seed-recovery system at the Coal Fired Flow Facility at the UTSI site.

1.3.2 Seed Regeneration Systems

Conversion of K_2SO_4 recovered from the exhaust gases into K_2CO_3 for purposes of recycling seed to the combustor can be accomplished by a number of chemical processes. The challenge to the MHD program is to select or develop a seed

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regeneration system which is economical in terms of capital equipment costs, operating costs and energy consumption. A process referred to as the "Formate" process was used in Europe for the production of K_2CO_3 . The Pittsburgh Energy Technology Center has proposed a more economical process for seed regeneration which also separates ash from the seed compounds. Testing has been performed on the bench-scale level. Further testing at larger scales is required in order to evaluate the economic feasibility of this scheme.

1.3.3 Characteristics of MHD Systems

Several comprehensive open-cycle MHD power plant parametric studies and conceptual designs have been conducted in the U.S. over the past five years. The purposes of these investigations have been primarily to assess the potential economic benefits to be gained from developing MHD technology, determine the optimum MHD power plant configurations for both early commercial and second generation technologies, and to establish a basis for planning and evaluating government-sponsored MHD research and development programs. These investigations have been funded by the Energy Research and Development Administration (ERDA), the Department of Energy (DOE) and the Electric Power Research Institute (EPRI).

The first of these comprehensive studies was the Energy Conversion Alternatives Study (ECAS), which was administered and managed by the National Aeronautics and Space Administration (NASA) - Lewis Research Center and supported by ERDA and the National Science Foundation (NSF). The purpose of this study was to develop estimates on power plant costs and energy production costs for a number of advanced coal-fueled electric energy conversion technologies including gas turbines, combined cycles, MHD, fuel cells, etc. The two major prime contractors were General Electric Co. and Westinghouse Electric Corporation. Avco-Everett Research Laboratory assisted GE with the MHD power plant analysis as a subcontractor. The ECAS program was conducted in two phases, beginning in 1975. In the first phase, both GE and Westinghouse examined several MHD power plant configurations with a number of variations in major design parameters. In the second phase, GE developed a conceptual design and analyzed costs and performance for the most promising MHD power plant configuration. MHD was not included by Westinghouse in the second phase.

Two investigations were conducted under EPRI sponsorship following the completion of ECAS. The emphasis of these investigations was directed towards several optional configurations not considered in ECAS and towards a technology base which was considered to be less advanced than that of ECAS. The two prime contractors in these investigations were Westinghouse and STD Research Corporation.

The most recent comprehensive comparative MHD study is the Parametric Study of Potential Barly Commercial MHD Power Generation (PSPEC). Contracts for the PSPEC program were awarded by NASA-Lewis Research Center to GE and Avco-Everett Research Laboratory and a grant was awarded to the University of Tennessee Space Institute (UTSI). As the name implies, the emphasis in the PSPEC program was on

"first generation" commercial MHD technology rather than on the more advanced technology status assumed in ECAS. Work on the PSEC MHD power plant concepts is being extended through a program called the Conceptual Design Study of Potential Early Commercial MHD Power Generation (CSPEC). NASA has awarded contracts for the CSPEC program to GE and Avco.

1.3.3.1 MHD Power Plant Design Variations

A large number of variations of MHD power plant configurations were considered in the ECAS, EPRI and PSPEC investigations. These variations are indicated in Table 1.3.3.1-1. In addition to varying the plant configurations, major plant parameters were also varied in order to determine the optimum plant designs and to determine the characteristics of each configuration. Parameters which were varied included:

- . air preheat temperature
- . oxygen enrichment level
- . ash carryover into channel
- . seed concentration in channel
- . combustor pressure and temperature
- . extent of coal drying
- . magnetic field strength
- . channel loading parameter

The plants ranged in size from approximately 600 MWe to 2000 MWe for the ECAS program and from approximately 400 to 1000 MWe for the PSPEC and the EPRI programs.

1.3.3.2 Descriptions of Selected MHD Cycles

Several of the configurations considered in the ECAS and PSPEC investigations will be described in this section because of the insights revealed and their relevance to the selection of an MHD power plant configuration to be considered for use with over-the-fence gasified coal and with an integrated coal gasifier.

Integrated Low-Btu Gasifier Cycle

The only cycle which included the use of a clean low Btu gas as a fuel for the MHD combustor was one which was investigated by Westinghouse in Phase 1 of the ECAS program, (Ref. 1.3-6). Figure 1.3.3.2-1 is a schematic diagram of this cycle. The gasifier is a Westinghouse fluidized bed gasifier incorporating high-temperature desulfurization. The gasifier is integrated into the MHD cycle, in that the gasifier air is bled from the MHD compressor discharge after being preheated by the MHD exhaust gas. The clean product fuel gas is also preheated by the MHD exhaust gas. Steam is also supplied to the gasifier from the steam bottoming plant. A schematic diagram of the Westinghouse fluidized bed gasifier is shown in Fig. 1.3.3.2-2 (Ref. 1.3-7).

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TABLE 1.3.3.1-1

VARIATIONS IN MHD POWER PLANT CONFIGURATIONS

Type of Fuel

Illinois No. 6 coal Montana Rosebud coal North Dakota lignite Solvent refined coal (SRC)

Types of Seed Material

Potassium Cesium

Fuel Utilization by MHD Combustor

Direct coal firing Integrated low Btu cleanup with fuel gas cleanup without fuel gas cleanup Char from air-preheater carbonizer

Combustor Type

Single stage Multiple stage

MHD Channel Interconnection

Faraday Diagnoal

Oxidizer

Preheated air (high temperature) Oxygen enriched air

Air Preheat Mode

Direct-fired high temperature air heater (DFHTAH) Sepa: ~elv-fired high temperature air heater (SFHTAH) atmospheric pressure pressurized

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TABLE 1.3.3.1-1 (Cont'd)

VARIATIONS IN MHD POWER PLANT CONFIGURATIONS

Fuel Source for SFHTAH

Direct coal firing with slagging combustor Vapor from coal carbonizer Integrated low Btu gasifier Chemically active pressurized fluidized bed

Oxygen Supply

Over-the-fence purchase On-site high-purity air separation plant On-site low-purity air separation plant

Bottoming Plant

Air turbine Subcritical steam plant with reheat Supercritical steam plant with reheat

Nitrogen Oxide Control Method

Combustion control with homogenous decomposition Supplemental ammonia injection Supplemental catalytic decomposition Scrubbing for maximum HNO₃ production

Sulfur Recovery Method

Seed regeneration Conventional flue gas desulfurization Low Btu Fuel Gas Desulfurization Combined sulfur recovery with HNO₃ production

Seed Regeneration Method

PETC (i.e., PERC or USBM) process Formate process

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FIG. 1.3.3.2-1



MHD POWER PLANT CYCLE WITH INTEGRATED LOW BTU GASIFIER

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FLUIDIZED BED GASIFICATION PROCESS WITH HIGH TEMPERATURE SULFUR AND PARTICULATE REMOVAL

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This cycle employs a direct-fired high temperature air heater. No oxygen is added to the air. The seed material is cesium. Since the sulfur removal is performed prior to combustion, the seed is recovered primarily in the form of cesium carbonate and recycled directly to the combustor along with additional seed to make up for losses. The steam bottoming plant employs a supercritical singlereheat steam cycle. For a nominal power plant output of 2000 MWe, the plant efficiency was estimated to be in the range of 48 to 54 percent, depending upon design conditions. This efficiency compares favorably with other ECAS configurations considered by Westinghouse, but the cost of electricity was estimated to be higher than those of the other configurations.

Direct Coal Fired Plant With Direct Air Preheating

The MHD cycle which was selected in Phase I of the ECAS program as the most economical cycle was a direct coal-fired MHD power plant with a direct-fired high temperature air heater. This is the cycle for which GE conducted a conceptual design study in Phase 2 of the ECAS program (Ref. 1.3-1). A simplified schematic diagram of the GE cycle is shown in Fig. 1.3.3.2-3.

In this cycle, the oxidizer is air which is preheated by the MHD exhaust gas. The seed is potassium, which is recovered from the exhaust gas primarily as K_2SO_4 and converted to K_2CO_3 for recycle to the combustor. The seed regeneration process is the "PERC" process which requires a synthesis gas such as that produced by a medium-Btu coal gasifier. The steam bottoming plant employs a supercritical single-reheat steam cycle. For a power plant output of 1932 MWe, the overall efficiency is estimated to be 48.3 percent. The gross MHD generator output is 1406 MWe. The gross steam plant electric power output is 587 MWe and the plant auxiliary electric load is 61 MWe. The compressor power requirement is 372 MW, which is provided by steam-turbine mechanical drive.

Figure 1.3.3.2-4 is a more detailed schematic diagram of this cycle. It illustrates the extent of integration of the MHD topping cycle, the MHD bottoming cycle and the seed regeneration system. Boiler feed water for the steam plant is heated in cooling passages in the walls of the combustor, MHD channel and diffuser; the air compressor is driven mechanically by a steam turbine; air for MHD combustion and air for seed reprocessing are preheated by the MHD exhaust gas which is also used for coal drying.

Direct Coal Fired Plant With Indirect Air Preheating

Both of the cycles described thus far in this section employ direct-fired high temperature air preheating. Although these cycles lead to high power plant efficiency, the direct-fired high temperature air heater is a component which requires significant advancement in the current state-of-the-art in order to attain commercial feasibility. It appears that the practical approach to early commercial MHD power generation is to employ a separately-fired high temperature air heater (SFHTAH) or to utilize a cycle which is based upon oxygen enrichment of the air.

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MHD POWER PLANT CYCLE WITH DIRECT-FIRED HIGH TEMPERATURE AIR PREHEATER

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Figure 1.3.3.2-5 is a schematic diagram of one of the SFHTAH cycles investigated by Avco in the PSPEC program (Ref. 1.3-2). The MHD combustor is directly fired with coal. However, the SFHTAH, which is more susceptible to plugging due to ash, is fired with a low Btu gas produced from coal in a series of Wellman-Galusha fixed bed gasifier units. All of the fuel gas produced in the gasifier is cleaned of particulate matter, but only part (approximately one half) undergoes sulfur removal (low temperature), as shown in Fig. 1.3.3.2-6. The fuel gas is preheated by MHD exhaust gas prior to humaning in the SFHTAH combustors. The products of combustion from the SFHTAH unives are mixed with the MHD exhaust gas stream prior to discharge to allow additional sulfur removal in conjunction with seed recovery. The MHD exhaust gas is also utilized for coal drying.

MHD power plants of this configuration were investigated in the size range of approximately 400 to 900 MWe net power output. Estimated efficiencies ranged from 41.4 to 43.2 percent, depending upon various design parameters. For a typical case with a net power output of 936.1 MWe, the MHD generator d.c. power output is 495 MWe and the steam plant output is 611.6 MW, including electrical power generated and mechanical steam drives for compressors and pumps. The compressor-drive power requirement is 113 MW and the total auxiliary load and related losses is 57.5 MW, resulting in a net plant power output of 936.1 MWe with an efficiency of 42.5 percent.

The seed regeneration method proposed by Avco for this cycle is the formate process. This process requires a gas rich in CO. The proposed source of CO is a coke gasifier. The formate process produces CaSO₄ which requires disposal. Heat from the MHD combustor channel and diffuser is used to heat boiler feedwater. The steam plant is a subcritical cycle with single reheat. 1

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Direct Coal-Fired Plant With Oxygen Enrichment

Figure 1.3.3.2-7 is a schematic diagram of the oxygen-enriched-air cycle investigated by Avco in the PSPEC program (Ref. 1.3-2). The MND combustor is directly fired with coal and the oxidizer is preheated to moderate temperature (within the current state-of-the-art) by the MND exhaust gas. The oxygen is provided by an on-site air separation plant which is integrated with the steam bottoming plant by utilizing steam-turbine drives for the air separation plant compressors. The air separation plant produces oxygen of 80-percent purity to provide a total oxygen content of the combustion gas of 29.2 to 39.1 percent by volume. The proposed seed regeneration method is the formate process utilizing a coke gasifier to provide the required CO. As in the previous cycle, CaSO₄ from the seed regeneration system must be discarded; MND combustor and channel cooling provide boiler feedwater heating; the air compressor is steam-turbine driven and coal drying is accomplished with MHD exhaust gas. The steam plant is a subcritical cycle with single reheat.

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MHD POWER PLANT CYCLE WITH SEPARATELY-FIRED HIGH TEMPERATURE AIR PREHEATER

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FIG. 1.3.3.2-6

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FIG. 1.3.3.2-7

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MHD power plants of this configuration were investigated in the size range of approximately 600 to 900 MWe net power output. Estimated efficiencies ranged from 42.9 to 44.4 percent, depending upon various design parameters. For a typical case, with a net power output of 930.7 MWe, the MHD generator d.c. power output is 523 MWe and the steam plant output is 636.3 MW including electrical power generated and mechanical steam drives for compressors and pumps. The compressor-drive power requirement is 175.5 MW and the total auxiliary load and related losses is 53.1 MW, resulting in a net plant power output of 930.7 MWe with an efficiency of 43.4 percent.

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1.3.3.3 Other MHD Configurations

A number of other investigations have been conducted in recent years which have a bearing on the selection of a configuration for the combination of a coal gasifer and an MHD power plant. The emphasis in the U.S. is on direct coal-fired MHD power generation. The program in the U.S.S.R. has been primarily directed towards the use of natural gas as a fuel. Although MHD experiments have been conducted in the U.S.S.R. with the combustion of coal, it is anticipated that the first Soviet large-scale MHD power plant will be fired with natural gas. The only countries which have recently conducted investigations involving low or medium Btu gas as a fuel for the MHD combustor are India, Poland and Sweden.

India has considerable reserves of coal with high ash content. The government of India is presently supporting an MHD research and development program which is aimed at developing an MHD plant using low Btu gas as a fuel. An investigation was conducted during the early stages of this program to compare the potential suitability of various gasification processes for MHD application (Ref. 1.3-8). The processes considered were Lurgi, blue water, Koppers-Totzek and producer gas. The primary criteria for comparison were the theoretical flame temperature and the electrical conductivity of the plasma under various conditions of air preheat temperature and oxygen enrichment. All of the gases except producer gas were determined to be acceptable, with Lurgi gas being the most promising. A pilot plant is presently under construction which will utilize blue water gas as a fuel because of its availability at the selected site.

The MHD research and development program in Poland is aimed at direct coal firing. Theoretical investigations have been conducted in Poland regarding MHD with coal gasification, but information on these investigations is not presently available. A study was recently conducted in Sweden of an MHD cycle in which coal is gasified by use of the exhaust gas from the MHD generator (Ref. 1.3-9). Figure 1.3.3.3-1 is a schematic diagram of this cycle. Investigations of this concept have previously been conducted in several countries, including the U.S. The concept leads to a very efficient power plant, but it is in a relatively primitive stage of development compared to conventional coal gasification techniques.


A major program in the U.S. is the Engineering Test Facility (ETF) program. This is a facility which is presently in the planning stage. The ETF is an MHD pilot plant with a net power output of several hundred MWe and is conceived to be the next major step between facilities now under construction and the first commercial scale MHD power plant. The ETF program was initiated with the award by ERDA of contracts to GE, Westinghouse and Avco for extensive ETF conceptual design studies. Upon completion of these studies, further work on MHD ETF design criteria and facility definition has been conducted under program management by NASA-Lewis Research Center. The importance of this work lies in the fact that decisions being made with regard to the MHD ETF configuration will largely determine the direction of current MHD research and development programs.

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2.3 Description of Advanced Combined MHD/Steam Power Plant

2.3.1 Establish Operational Characteristics

As indicated in the equations presented in the previous section, it is necessary for the plasma to have a high electrical conductivity, a high velocity and the strength of the imposed magnetic field must be high in order to extract electrical power economically. The desired order of magnitude of the electrical conductivity is approximately 10 ohm⁻¹ m⁻¹; the Mach number is typically in the high subsonic range and the magnetic field strength should be in the range of 5 to 7 tesla (where 1 tesla is equal to 10,000 gauss).

Figure 2.3.1-1 indicates the means by which the desired levels of conductivity, velocity and magnetic field strength are achieved. A gas becomes electrically conducting when free electrons are present in the gas. The free electrons are introduced into the gas when one or more of the constituents of the gas become ionized. This requires elevating the gas to a high temperature and introducing a "seed" material (a compound containing potassium or cesium) which can more readily be ionized at temperatures which can be achieved by practical means. The high temperature is achieved by combustion of a fuel and air. However, the combustion temperatures, which are attainable from combustion of fossil fuels in air are typically in the vicinity of 2200°K (3500°F) which is inadequate for obtaining the desired electrical conductivity. In order to achieve a temperature in the vicinity of 2800°K (4600°F) which is desired for MHD, the combustion air must be preheated or the oxygen content of the air must be increased. Figure 2.3.1-2 from Ref. 2.3-1 shows the effects of air preheat temperature and oxygen enrichment on combustion temperature.

The desired velocity in the MHD channel corresponds to Mach numbers ranging from slightly subsonic to slightly supersonic. The velocity level is readily achieved by compressing the combustion air to pressures in the range of 5 to 10 atmospheres. Expansion of the combustion products through a subsonic or supersonic nozzle connecting the combustion chamber to the MHD channel provides the required velocity.

Magnetic field strengths as high as 5 to 7 tesla are attainable by the use of a superconducting magnet. This type of magnet consists of coils of special materials which completely lose their electrical resistivity when cooled to a very low temperature. Superconducting magnet systems for MHD operate at the temperature of liquid helium at atmospheric pressure which is 4.2°K. The only power requirements for the magnet are for initial energizing of the coils and for refrigeration to keep the temperature at the desired level.



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2.3.2 Basic MHD Power Plant Configurations

Based upon the requirements for obtaining the desired levels of electrical conductivity, velocity and magnetic field strength, two basic minimum MHD power generating plant configurations are shown in Fig. 2.3.2-1. One utilizes oxygen enrichment and the other utilizes an air preheater. Typical preheat requirements are 1650 to 1900°K (2500 to 3000°F) with no oxygen enrichment or 850°K (1100°F) with an overall oxygen concentration of 30 to 40 percent by volume. In either case, the oxidants are fed under compression to the combustor along with fuel and a seed material. The conducting gases which flow through the MHD channel consist of the products of combustion and the seed material. The seed material is injected into the combustor as a potassium or cesium compound such as K_2CO_3 (potassium carbonate). At the high temperature attained in the combustor, the seed compound vaporizes and potassium ions are formed, providing free electrons.

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The temperature of the gas decreases as it passes through the channel due to extraction of energy in the form of electricity. When the gas temperature decreases to approximately 2300° K (3700° F), the conductivity is too low to produce further electrical power economically. This limits the electric power which can be extracted from the MHD generator. The MHD generator configurations shown, however, can be used as topping units in conjunction with a steam bottoming cycle.

The electrical power output of the MHD channel is in the form of direct current. This may be suitable for certain industrial applications such as aluminum production. However, for electric utility applications feeding into an existing grid system, an inverter system is required for changing the direct current output to alternating current, as shown in Fig. 2.3.2-1.

One means of attaining a higher efficiency is by utilizing the high temperature exhaust gas from the MHD generator channel to preheat the air prior to combustion as shown in Fig. 2.3.2-2. This is referred to as direct-fired air preheating. However, the technology of achieving such high air temperatures with a gas which contains seed materials is beyond the present state of the art. At the present time, an upper limit of air temperature for direct-fired preheating is in the 1100 to 1300°K (1500 to 1900°F) temperature range.

Full utilization can be made of the MHD generator exhaust gas by adding a steam bottoming plant to the system discussed previously. A simple version of such a combined MHD/stram power plant cycle is shown in Fig. 2.3.2-3. The heat from the gas leaving the air preheater produces steam which drives a steam turbine connected to a conventional electric generator. The a.c. output of this generator is combined with the a.c. output of the MHD generator inverter system to provide power to an electric utility network or to an industrial plant. For air preheat temperatures in the vicinity of 2500°F, an overall power plant efficiency of nearly 50 percent is projected.

FIG. 2.3.2-1





BASIC MHD POWER GENERATION SYSTEMS UTILIZING (A) OXYGEN ENRICHMENT AND (B) SEPARATELY-FIRED AIR PREHEATING

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BASIC MHD POWER GENERATION SYSTEM UTILIZING DIRECT-FIRED AIR PREHEATING



Figure 2.3.2-4 shows the combined MHD/steam cycle with the addition of systems for cleaning the exhaust gas prior to emission from the stack and for processing the seed for recyling to the combustor. The gas clean-up system consists of a particulate removal system such as a bag filter system or an electrostatic precipitator which is capable of collecting fine particles. This serves the dual prupose or recovering the seed material and reducing pollutant emission levels. By the time the exhaust gas has reached the gas clean-up system, it has been cooled to the point at which seed compounds have first condensed and then solidifed and formed primarily K2CO3 and K2SO4. There is a strong tendency for the formation of K₂SO₄ which limits the amount of sulfur oxides which enter the atmosphere without the need for a flue-gas desulfurization system. The gas clean-up statem removes both flyash and seed compounds from the gas stream. If the fuel contains negligible sulfur, the collected seed compound would be primarily K_2CO_3 which can be recycled directly to the combustor. If the fuel does contain a significant amount of sulfur, the seed compounds include K_2SO_4 which must be converted into K_2CO_3 before recycling to the combustor in order to meet environmental regulations on stack emissions of sulfur oxides. This conversion is accomplished in a chemical processing system which may be located at the plant site.

2.3.3 MHD Power Plant Components and Subsystems

The MRD power plant can be considered to be subdivided into three categories:

- . MHD Power Train
- . Heat Recovery Seed Recovery System
- . Balance of Plant

The MHD power train includes the combustor, channel, diffuser, magnet and inverter. The channel has been described previously. The combustor, magnet and inverter are described in later sections. The diffuser is a diverging duct located at the downstream end of the channel. Its function is to efficiently reduce the velocity of the gas before it enters the remainder of the plant.

The heat recovery seed recovery (HRSR) system includes the equipment through which the gas flows after leaving the diffuser. The functions of the HRSR are to cool down the gas while transferring heat from the gas to the boiler feedwater, steam and air and to remove seed compounds and ash from the gas. All other components and subsystems may be classified as balance-of-plant equipment. These include the steam turbine-generator, compressors, pumps, fans, piping, electrical equipment, etc., as well as equipment which may be unique to an MHD power plant such as the seed regeneration system.

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2.3.3.1 Coal Processing and Combustion

There are several proposed approaches to the utilization of coal as the fuel for the MHD generator. Three approaches are illustrated schematically in Fig. 2.3.3.1-1. As indicated previously, the MHD generator can be designed to operate in the presence of coal slag. Thus, the direct-firing of coal in the combustor is a viable option. However, the slag has been observed to absorb seed material when the slag and seed cool down as the as flows through the components downstream of the MHD generator. The seed absorbed by the slag is very difficult to recover. The amount of seed absorbed is found to be unacceptable unless approximately ninety percent of the slag is removed either in the combustor, as in Fig. 2.3.3.1-1(a) or immediately downstream of the diffuser as in Fig. 2.3.3.1-1(b). The third method of coal utilization is to gasify the coal prior to combustion as indicated in Fig. 2.3.3.1-1(c). This can be done with or without fuel-gas sulfur removal since sulfur removal can be accomplished downstream of the MHD generator in conjunction with seed recovery and regeneration.

Several combustor designs have been developed for MHD systems which are capable of extracting large amounts of ash from the combustion products. One method is to use a two-stage combustor in which the temperature of combustion in the first stage is controlled by delivering less air to it. The slag is removed from the first stage and the remainder of the air is added to the second stage. Removal of over 90 percent of the slag from a two-stage combustor has recently been demonstrated at the U.S. Department of Energy's Pittsburgh Energy Technology Center with a two-stage combustor as illustrated in Fig. 2.3.3.1-2. Another two-stage combustor design which has been proposed by TRW is shown in Fig. 2.3.3.1-3. A single-stage combustor, designed by Avco-Everett Research Laboratory for achieving high slag removal, is shown in Fig. 2.3.3.1-4.

2.3.3.2 Systems for Control of Pollutant Emissions

Nitrogen Oxides

The very high temperatures occurring in the combustor of an MHD power plant are conducive to the production of nitrogen oxides through the fixation of nitrogen which is present in both the air and the fuel. Unless precautions are taken in the design of the plant, NO_x emissions will exceed allowable environmental limits. Figure 2.3.3.2-1 illustrates schematically the design features which are instituted to reduce NO_x emissions to acceptable levels. The amount of air delivered to the combustor is maintained at approximately 95 percent of the stoichiometric requirements in order to reduce the initial production of nitrogen oxides. However, the incomplete combustion which thereby occurs results in production of excessive amounts of carbon monoxide. This makes it necessary to add additional air at a location where the gas has cooled to the point that the oxidation of nitrogen is no longer significant. This should be done in the 2400°F to 2900°F

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(C) CLEAN FUEL FROM COAL GASIFIER

SLAG REMOVAL SCHEMES IN COAL-FIRED MHD POWER PLANTS

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FIG. 2.3.3.1-2



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TWO-STAGE DIRECT COAL-FIRED MHD COMBUSTOR DESIGN



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temperature range. Additional decomposition of NO_x is achieved by control of the cooling rate of the gas as it passes through the critical temperature range of 3300°F to 2900°F. This is achieved by ensuring sufficient residence time in the components immediately downstream of the diffuser. For example, a radiant boiler at this location would provide the required cooling rate for the gas.

Particulates

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The particulate matter in the exhaust gas includes ash and compounds of potassium. Typical potassium compounds may incude:

K₂SO₄ K₂CO₃ KOH KO KHCO₃ K₂S K₂S₂O₇ KHSO₄

For fuels with high sulfur content, the predominant compound at low temperatures is K_2SO_4 . For fuels with low sulfur content, the predominant compound at low temperatures is K_2CO_3 .

Condensed or solidified materials adhere to surfaces of components in the equipment downstream of the generator or are entrained in the gas stream and carried further downstream. Liquid or solid materials are continually or intermittently removed from the surface (for example, by sootblowers) to prevent excessive build-up and seed loss. The materials removed from the surfaces are either collected at the bottoms of the various components or entrained in the gas stream. Particles which are carried in the gas stream are collected by either a baghouse filter or an electrostatic precipitator to prevent excessive pollutant emissions as well as for the purpose of recovering seed.

Sulfur Oxides

As discussed previously, the mechanism most frequently proposed for sulfur emissions control in an MHD power plant is by means of the formation of K_2SO_4 and its subsequent removal from the MHD exhaust stream by means of baghouse filters or electrostatic precipitators. The potassium is required for MHD to enhance the electrical conductivity of the gas. For fuels with moderate-to-low sulfur content, the amount of seed required for achieving adequate gas conductivity is more than is required to extract sulfur from the gas. For fuels with high sulfur content, the potassium which must be added for sulfur extraction may be

only slightly greater than that required for conductivity. Due to the cost of potassium, it is necessary to recover as much of it as possible to re-use in the combustor. Thus there is no added penalty associated with the removal of sulfur oxides from the gas stream. However, the recovered K_2SO_4 must be converted to K_2CO_3 before recycling. It has been projected that approximately 95 percent of the potassium must be recovered for economic reasons.

Seed Regeneration

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The amount of seed required to achieve adequate electrical conductivity of the ionized gas is equal to approximately 15 percent of the mass of fuel when the fuel is coal. It is estimated that at least 95 percnet of this seed must be recovered and recylced to make the plant operating cost acceptable. When low sulfur fuels are used, the seed is collected primarily as K_2CO_3 with small amounts of K_2SO_4 . Sufficient K_2SO_4 must be converted to K_2CO_3 before recycling to the combustor to meet environmental regulations. When high sulfur fuels are used, most of the collected seed is in the form of K_2SO_4 and a large percentage of it must be converted to K_2CO_3 before recycling. The regeneration process includes separation of ash from the seed as well as conversion of potassium sulfate to potassium carbonate.

A seed regeneration process which has been investigated at the Pittsburgh Energy Technology Center involves two stages with the following reactions:

Reduction:

 $K_2 SO_4 + 4 CO \rightarrow K_2 S + 4 CO_2$ $K_2 SO_4 + 4 H_2 \rightarrow K_2 S + 4 H_2 O$

Regeneration:

 $K_2S + CO_2 + H_2O \rightarrow K_2CO_3 + H_2S$

In this process, coal is supplied as a fuel to provide a $CO-H_2$ producer gas for the process and the hydrogen sulfide is converted to elemental sulfur in a Claus plant. This process is presently in the "bench-scale" stage of development.

An alternate regeneration process is the Formate process based upon the chemical reaction

 $K_2SO_4 + Ca (OH)_2 + 2 CO \rightarrow 2 KCOOH + CaSO_4$

This process was used commercially for production of K_2CO_3 from K_2SO_4 , but has since been abandoned as it became more economical to produce K_2CO_3 from other feedstocks. In an MHD power plant, the steps in the process would be:

- 1. Gasification to produce carbon monoxide (CO);
- 2. Dissolving of recovered seed (K_2SO_4) and slurrying of lime $(Ca(OE)_2)$;
- 3. Reaction of CO, K_2SO_4 and $Ca(OH)_2$ to produce soluble potassium formate (KCOOH) and insoluble calcium sulfate (CaSO₄);
- 4. Filtration of CaSo₄;
- 5. Evaporation of water from KCOOH;
- 6. Recycle of KCOOH to combustor.

2.3.3.3 Heat Recovery-Seed Recovery System

There are a number of alternate ways in which the heat recovery components downstream of the diffuser can be arranged. The heat recovery-seed recovery (HRSR) components include steam generating equipment, superheater, reheater, air preheater(s), economizer(s), a bag-house filter or electrostatic precipitator, an induced-draft fan and a stack. A portion of the gas stream may be diverted for the purpose of drying the coal. Figure 2.3.3.3-1 shows one possible arrangement.

The system depicted in Fig. 2.3.3.2-1 employs a direct-fired high-temperature air heater. The first component through which the MHD generator diffuser exhaust gas flows is a radiant boiler. This component is placed here primarily because of the need for high gas residence time to allow slow cooling through a critical temperature range to enhance the decomposition of nitrogen oxides. The secondary air for completion of combustion may be added before or after the gas flows through the high-temperature air heater depending upon the temperature drop of the gas through the radiant boiler section. The steam superheater and steam reheater sections of the plant follow the high-temperature air heater.

The reason for splitting the air heater into a high-temperature and a lowtemperature section is due to materials considerations. The low-temperature air heater would be a recuperative type of heat exchanger of metallic construction. The upper limit of this type of heat exchanger would be an air outlet temperature in the vicinity of 1100° K (1500° F) due to the presence of seed materials in the gas. The high-temperature air preheater would be of the regenerative type. The outlet temperature of the air from the high-temperature air heater is highly dependent upon advances in the state of the art. Figure 2.3.3.3-1 also shows an economizer located downstream of the low-temperature air heater. The economizer brings the gas temperature down to a range which is suitable for an electrostatic precipitator or a bag filter system.



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FIG. 2.3.3.3--1

Several alternate heat recovery system flow train arrangments are shown in Fig. 2.3.3.3-2. Each is based upon a different means of achieving a high air preheat temperature. Arrangement A shows a direct fired air preheater. Arrangements B, C and D show separately-fired arrangements. Arrangement B utilizes clean fuel fired at atmospheric pressure. Arrangement C utilizes clean fuel fired at high pressure with the combustion gases driving a turbine after preheating the air. Arrangement D utilizes low-Btu gas provided by a coal devolatilizer. 1

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Figure 2.3.3.3-3 is a schematic diagram of the condensate, feedwater and steam flow system in the steam bottoming plant. A single dearrating feedwater heater is shown. Additional feedwater heaters may or may not be necessary due to the substantial thermal energy which can be picked up from the MHD exhaust gas and the cooling system for the combustor, MHD channel and the diffuser. The cooling of the MHD flow train is achieved by means of secondary flow loops. The air compressor is steam-turbine driven. This requires a substantial amount of power. For example, a plant with a net electrical power output of 1000 MW may utilize as much as 150 MW of mechanical drive power for the air compressor systems. The main steam turbine shown is a single reheat turbine.

2.3.3.4 High-Temperature Air Preheating

The highest air preheat temperatures can be achieved by the intermittentlyoperating regenerative type of air preheater. This type consists of several vessels, each encasing a matrix of ceramic refractory materials. The matirx is alternately heated by the hot gas and cooled by the air, thereby tranferring heat from the gas to the air. Multiple vessels in parallel and switchover values are employed to deliver a continuous supply of preheated air. Figures 2.3.3.4-1 and -2 are simplified diagrams showlig a vessel cross-section with a matrix consisting of hexagonal-shaped cored bricks with internal flow passages. As many as 10 to 20 vessels may be required. Maximum air preheat temperatures in excess of 3000°F have been achieved with minimum deterioration of matrix materials utilizing a clean gas as a heat source. Air preheaters utilizing, as a heat source, gases containing slag or seed are not yet capable of sustained operation at temperatures high enough for economical operation of an MHD/steam power plant.

2.3.3.5 Superconducting Magnet

Magnetic field strengths of 5 to 7 teslas are required over volumes of several hundred cubic meters in order to obtain the overall efficiency advantages of MHD power generation. These field strengths are much greater than the 2-tesla field strengths that can be provided by a water-cooled conductor, iron pole magnet of medium power. The power requirements of a 6-tesla magnet using water-cooled conductors would exceed the power output of the generator. Therefore, large-volume superconducting magnets are required for efficienct MHD power generation. The only power continuously consumed by a superconducting-magnet system is that which is required for refrigeration to maintain the magnet coils at liquid helium temperatures.

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DIAGRAM OF REGENERATIVE HTAH SYSTEM

FIG. 2.3.3.4-1

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SECTION OF REGENERATIVE HIGH TEMPERATURE AIR HEATER VESSEL AND CORED BRICK

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The superconducting material being considered for MHD application is niobiumtitanium (NbTi) which is a superconductor at the temperature of liquid helium (4.2°K). In order to maintain the material in a superconducting state it is necessary to "stabilize" the superconductor. Stability is achieved by constructing the coils with NbTi imbedded in a copper substrate. This prevents the entire coil from reverting to its "normal" (non-superconducting) state when small local temperature excursions cause a small region of the coil to temporarily go normal.

The most efficient configuration for an MHD magnet is the "saddle coil" illustrated schematically in Fig. 2.3.3.5-1 from Ref. 2.3-2. The coils must be cooled by circulation by liquid helium and the entire magent structure must be thermally insulated from the channel in the bore of the magnet as well as from the surroundings. The extremely high electric currents and magnetic field strengths existing in the magnet generate very large stresses which tend to distort the magnet. A very substantial structure is required to restrain these forces. Figure 2.3.3.5-2 from Ref. 2.3-3 illustrates a superconducting magnet which is currently under construction or in the initial stages of construction for the Component Development and Integration Facility. Figure 2.3.3.5-3 from Ref. 2.3-4 is a block diagram of the cryogenic support system required for maintaining the magnet coils at liquid-helium temperature.

2.3.3.6 Inverter Systems

The MHD generator produces direct current electricity which must be converted to alternating current for electric utility utilization. The inverter system which converts d.c. power to a.c. power must meet the unique requirements of an MHD generator. For the Faraday type of MHD channel configuration the d.c. electric power is extracted from the channel via a very large number of individual electrode parts. It is necessary to consolidate these individual outputs in conjuction with the inverter system. Figure 2.3.3.6-1 from Ref. 2.3-1 illustrates a consolidation network for combining the electrical output of twelve individual electrodes. Figure 2.3.3.6-2 also from Ref. 2.3-1 illustrates, schematically, the manner of consolidating the outputs for either a Faraday or Diagonal type of channel. Although the inverter network for MHD application is unique, the equipment required for the inverter system is not unique for MHD.

2.3.4 Recommendations for Selection of MHD Power Plant Baseline Technology

Several alternate MHD power plant configurations were identified and described in the preceding sections of this report. The selection of one or more of these configurations depends upon the guidelines and selection criteria which are established. The major guidelines followed in selecting an MHD power plant configuration are that the MHD power plant be of a type which is:

. suitable for "early commercial" operation by a utility company

. economically attractive compared to alternate configurations

. currently receiving support under the U.S. Government MHD program.

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MHD MAGNET SADDLE COIL CONFIGURATIONS

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FIG. 2.3.3.5-2



Interior Arrangement, CDIF Superconducting Magnet

SUPERCONDUCTING MAGNET DESIGN FOR MHD COMPONENT DEVELOPMENT AND INTEGRATION FACILITY

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SCHEMATIC DIAGRAM OF CRYOGENIC SUPPORT SYSTEM FOR SUPERCONDUCTING MAGNET

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FIG. 2.3.3.8-2



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CONSOLIDATION NETWORK FOR 12 ELECTRODES

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CONSOLIDATION SCHEMES FOR FARADAY AND DIAGONAL CHANNELS

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The first guideline narrows the choice to cycles which employ separatelyfired high temperature air heaters or oxygen enrichment since directly-fired high temperature air heaters require considerably more development. The second criterion favors the oxygen enrichment cycle since it appears to have a slight economic edge over the separately-fired high temperature air heater case. The third criterion also favors the oxygen enrichment case since a "decision was made by the Director of the DOE Office of MHD in August 1979 to select the oxygen enrichment option for use in the ETF and the target commercial plant," according to a recent report (Ref. 2.3-5).

On this basis, the oxygen enrichment option has been selected for the present investigation and the specific configuration and design parameters of the oxygen enriched cycle investigated by Avco in the PSPEC program have been selected as a reference plant. The baseline technology characteristics are given in Table 2.3.4-1.

TABLE 2.3.4-1

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BASELINE TECHNOLOGY FOR COAL-FIRED MHD COMBINED CYCLE POWER PLANT

Illinois No. 6 Coal

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Oxidizer Flow, Combustion Temperature, Pressure Ratio Magnet Strength, Power Output,	lb/sec (34% O ₂) F Tesla MW(dc)	1008 4675 9.1 6 554			
			Steam System		
			Throttle/Reheat Temperature,	F/F	1000/1000
			Throttle/Reheat Pressure,	psi/psi	2400/444
			Throttle/Reheat Flow,	lb/sec/lb/sec	974/735
Power Output	MW	684			
Overall Plant					
Total Bower	NU	1238			

Total Power,	MN	1238
Net Power,	MW	978.5
Heat Rate,	Btu/kWh	7828

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3.3 Evaluation of Coal-Derived Fuels for MHD Power Generation Systems

The objective of this task is to determine the overall power plant efficiency of an open-cycle MHD power plant using medium-Btu MBa gas delivered to the power plant via pipline. The properties of the gas delivered to the power plant will be established and any further processing, if necessary, will be identified. Based upon the fuel gas characteristics, the performance of the MHD power plant will be described.

The major emphasis in the Department of Energy MHD program is on direct coal-fired (DCF) open-cycle MHD power generation. This emphasis is based upon the results of a number of comparative investigations in which direct coal-firing has been judged to be the most economical approach. As a result, a large amount of data has been developed on cycle performance and plant costs for DCF MHD. The only comprehensive investigation currently being conducted in MHD power generation utilizing coal-derived gas as a fuel is in India (Ref. 3.3-1).

In the investigation reported herein, a DCF MHD plant is selected as a reference case. Modifications due to using MBG as a fuel are identified and the changes in performance are calculated. This procedure takes advantage of the considerable amount of recently-developed estimates of DCF MHD power plant performance (Ref. 3.3-2).

3.3.1 Definition of Fuel Gas Requirements

One of the most critical parameters of an MHD generator is the electrical conductivity of the plasma. The plasma acquires its electrical conductivity as the seed material ionizes at very high temperature. For a given amount of seed material in the products of combustion, the electrical conductivity is very sensitive to the temperature, so that special measures are taken to achieve high flame temperature. These measures include enrichment of the combustion air with oxygen and preheating the enriched air and, possibly, the fuel. For a given fuel and a given oxidizer (enriched air), the conductivity at a given temperature depends strongly on the composition of the products of combustion.

A complete determination of the suitability of a coal-derived gaseous fuel for application in an MHD power plant and the optimum fuel conditions requires the determination of the electrical conductivity of the products of combustion over a range of fuel and oxidizer conditions. Because an accurate determination of the electrical conductivity of candidate fuels is beyond the scope of the present investigation. The approach taken herein is to determine the conditions which are necessary for attaining the same levels of electrical conductivity as are projected for the DCF MHD power plant reference case.
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The oxygen-blown Texaco and British Gas Corporation (BGC) slagging gasifiers were selected as the gasifiers to provide medium-Btu fuel gas for transport in pipelines to the power generation site. Properties of the fuel gas from each of these gasifiers, based on Ref. 3.3-3, are given in Table 3.3.1-1. Since the conductivity of the products of combustion were not determined directly, it was ascertained whether or not the flame temperatures are sufficiently high for achieving adequate conductivity. Thus flame temperatures were determined for combustion of the two types of fuel gas with several different oxidants under several sets of conditions. The results of the flame temperature calculations are given in Table 3.3.1-2. Oxidizer compositions investigated include unenriched air (20.74 percent 0_2 by volume) and enriched air with oxygen concentrations of 28 percent and 35 percent. In all cases, the oxidizer was considered to be preheated to 1100°F. The fuel gas was considered to be preheated to 1100°F or to be delivered to the combustor at the pipeline temperature of 68°F. The pressures considered were 6 atmospheres and 8 atmospheres. 1

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As seen in Table 3.3.1-2, the flame temperatures determined range from a minimum of 4207°F for the combustion of the Texaco fuel gas with unenriched air at 8 atmospheres to a maximum of 4870°F for the combustion, at a ;ressure of 8 atmospheres, of the BGC slagging gasifier fuel gas with air enriched to 35 percent 0_2 by volume with both the oxidizer and the fuel gas preheated to 1100°F. The acutal stagnation temperature at the inlet to the MHD generator will be less than the estimated flame temperature due to heat loss to the walls of the combustor and heat absorption due to vaporization and ionization of the seed.

In the DCF MHD reference plant, the MHD generator inlet stagnation temperature is in the 4650 to 4800°F range. Achieving an MHD generator inlet stagnation temperature in this range for the MBG-fired MHD plant is estimated to require an adiabatic flame temperature in the 4750°F to 4900°F range. It is observed in Table 3.3.1-2 that the lower end of this range can be achieved with the Texaco MBG at a pressure of 8 atmospheres utilizing air enriched with oxygen to 35 percent by volume if the oxidizer and the MBG are preheated to 1100°F. Under the same conditions, the adiabatic flame temperature for combustion of BGC medium-Btu gas approaches the upper end of this range. The MRG is delivated to the plant site at pressures well in excess of the required pressure and at a temperature above 75°F to prevent condensation of moisture. No further conditioning of the fuel gas is required except for preheating to achieve the necessary flame temperature. The exhaust gas from the MHD generator will be utilized to preheat the MBG as well as the oxidizer.

TABLE 3.3.1-1

GAS COMPOSITIONS

Heating Values, LHV

LHV,	Btu/scf	267.5	349.1
LHV,	Btu/1b	4949	6825

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ADIABATIC FLAME TEMPERATURES FOR COMBUSTION OF MEDIUM-BTU GASES

Fuel Gas Type	Oxidizer Oxygen Content, Volž	Combustion Pressure, atm	Oxidizer Preheat Temperature, ^o F	Fuel Gas Preheat Temperature, ^O F	Adiabatic Flame Temperature, ^O F
Техасо	20.74	œ	1100	1100	4207
Texaco	28.00	Q	1100	68	4402
Texaco	28.00	6	1100	1100	4539
Texaco	28.00	80	1100	68	46.44
Texaco	28.00	œ	1100	1100	4575
Texaco	35.00	9	1100	68	4578
Texaco	35.00	9	1100	1100	4710
Texaco	35.00	ø	1100	68	4616
Texaco	35.00	œ	1100	1100	4753
BGC	20.74	80	1100	1100	4344
BGC	28.00	9	1100	68	4522
BGC	28.00	9	1100	1100	4630
BGC	28.00	80	1100	68	4555
BGC	28.00	œ	1100	1100	4668
BCC	35.00	Q	1100	68	4719
BGC	35.00	9	1100	1100	4825
BGC	35.00	œ	1100	68	4760
BGC	35.00	œ	1100	1100	4870

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3.3.2 Identification of Power Generation Facility

The direct coal-fired MHD power plant configuration which has been selected as a reference plant is the conceptual design for an oxygen-enriched MHD power plant investigated by Avco-Everett Research Laboratory in the Parametric Study of Potential Early Commercial (PSPEC) MHD Power Plants (Ref. 3.3-2). A schematic diagram of the cycle configuration for this reference plant is shown in Fig. 3.3.2-1. A schematic diagram of its steam and feedwater cycle configuration is shown in Fig. 3.3.2-2.

This reference plant configuration was investigated in the PSPEC MHD study over a range of sizes and parameters. The base case is a 931-MWe plant utilizing enriched air with 35 percent oxygen by volume preheated to 1100°F by the MHD generator exhaust gas. The fuel is Montana Rosebud coal. Variations investigated included power plant size reduction to 614 MWe, oxygen enrichment levels ranging from 29 percent to 39 percent, oxidizer preheat levels ranging from 1000°F to 1400°F and utilization of Illinois No. 6 coal as an alternate fuel.

The oxygen for enriching the air is provided on an on-site air separation plant which produces oxygen with a purity of 80 percent. The air separation plant compressor-drive turbines are driven by steam from the MHD power plant bottoming cycle. Steam from the bottoming cycle also drives the compressors for the combustion air. The steam bottoming plant boiler feedwater is used for cooling the combustor, MHD generator and diffuser. The steam turbine is a subcritical unit operating at 2400 psia/1000°F inlet conditions with 1000°F single reheat. Stack gas from the steam bottoming plant is used for drying the coal. Extensive coal drying is required in order to maximize the flame temperature and the plasma electrical conductivity. The recovered seed is converted from K_2SO_4 to K_2CO_3 by means of the formate seed regeneration process which produces $CaSO_4$ as a by-product which requires disposal. The formate process also requires fuel gas produced from an on-site coal gasifier.

The performance of the MBG-fired MHD plant is determined by identifying the differences between the MBG-fired MHD power plant and the reference plant. These differences arise from the fuel and fuel characteristics. For the DCF MHD plant, there is need for coal handling and processing and means must be provided for dealing with the coal ash and sulfur. The MBG is essentially free of ash and sulfur and has different chemical composition and heating value. The differences between the two types of MHD power plant may be categorized as differences in plant configuration, operational differences and performance differences.

The change in fuel means the elimination of facilities and equipment for delivery, storage drying, pulverizing, pressurization and injection of coal. These



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STEAM AND FEED WATER CYCLE CONFIGURATION FOR DIRECT COAL-FIRED MHD REFERENCE POWER PLANT

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items are replaced with medium-Btu gas piping and flow controls. An additional heat exhanger is required for preheating the MBG prior to combustion.

The absence of coal ash in the MBG fuel has several implications. The combustor is considerably simplified since it is not necessary to remove slag from the combustor. There will be no build-up of a slag layer on the walls of the MHD generator. This entails a considerable change in MHD channel design which is related more to channel lifetime and materials selection than to channel performance. The equipment downstream of the MHD generator will be relieved from problems associated with slag, including deposits, corrosion, fouling, seed absorption and ash removal. There is also no requirement for facilities and equipment for ash handling and disposal.

The absence of sulfur in the MBG fuel introduces the possibility of several modifications in MHD power plant configuration.^{*} Since the seed does not need to play a role in sulfur emissions control, the amount of seed injected into the combustor is the optimum amount required for achieving high plasma electrical conductivity, rather than the amount required to achieve the necessary sulfur emissions control. Furthermore, the option becomes available of injecting the seed into the combustor in the form of K_2SO_4 instead of a sulfur-free compound. The need for a seed regeneration system is eliminated since the fuel is virtually sulfur-free. This not only eliminates the seed regeneration facility itself, but also the associated coal gasifier, the CaO delivery and handling system and the CaSO₄ disposal system.

The changes in configuration of the heat recovery seed recovery system (HRSR) include the addition of a heat exchanger to transfer heat from the MHD generator exhaust gas to the MBG fuel and the elimination of the need to divert some of the stack gas to the coal drying system. The elimination of design constraints due to the absence of ash and sulfur in the gas stream introduce additional flexibility in certain design details which may lead to small improvements in performance and reductions in cost. A major change in balance-of-plant systems arise from elimination of ash and CaSO₄ disposal. Figure 3.3.2-3 is a schematic diagram of the cycle configuration for the MBG-fired MHD plant based upon the cited modifications of the DCF MHD plant.

The overall performance of each of the MBG-fired MHD power plants is determined by estimating the performance modifications for each component and subsystem and

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[&]quot; In the DCF MHD power plant, sulfur is captured by the seed material and collected from the gas stream in the process of recovering the seed. Sulfur is removed external to the cycle in a seed regeneration system and the seed is recycled to the combustor in a sulfur-free form.





FIG. 3.3.2-3

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determining their cumulative effects. The component and subsystem performance modifications are determined by making direct comparisons between the MBG-fired MHD power plants and the DCF MHD power plant. The thermal input to the MBG-fired MHD power plants is taken to be the same as the thermal input to the MHD combustor of the reference plant, based upon the fuel higher heating value. This input is 2174 Mwt (7420 x 10^6 Btu/hr).

The operating conditions such as magnetic field strength and steam throttle conditions for the MBG-fired MHD power plants are generally taken to be the same as for the DCF MHD reference plant. However, combustor pressures for the MBG cases are taken to be greater than that for the DCF case. The reason for this is that the selection of the pressure is based upon a trade-off between MHD generator output and the power consumption of the oxidizer compressor, both of which increase with increasing pressure. Since the oxidizer flow rates for the MBG cases are less than that for the DCF case, it is advantageous to operate at higher pressure. This helps to compensate for the lower combustion temperatures of the plasma for the Texaco MBG case.

The MHD generator output will be lower for the Texaco MBG case, due primarily to the lower combustion temperature, and, hence, lower conductivity. The heat loss for the MBG-fired MHD combustors are less than that of the DCF MHD because of the requirement for slag removal from the combustion products for the DCF combustor. The MHD channel and diffuser heat loss for the DCF is less than that for the MBG cases because of the insulating effect of the slag layer in the DCF case. However, the heat losses from the combustor, channel and diffuser are picked up by boiler feedwater for the steam bottoming plant.

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The compressor power consumption for the oxidizer for the MBG cases is reduced relative to that for the DCF case due to the lower oxidizer flow rates for the MBG-fired power plants. The air separation plant power consumption for the MBG cases is also lower than that for the DCF case because of the reduction in oxygen demand.

The oxidizer preheating requirement is also reduced for the MBG cases because of the lower oxidizer flow rates. However, this advantage is more than offset by the necessity to preheat the medium-Btu gas itself in order to achieve high combustion temperatures. The requirements for coal drying energy and fuel for seed reprocessing are eliminated for the MBG cases.

The estimated performance for the two MBGF MHD power plants are given in Table 3.3.2-1. The performance levels for the Texaco and BGC gas fuels are 43.4 and 44.5 percent, respectively, based upon fuel higher heating value. The efficiencies based upon lower heating values are listed, respectively, as 46.3 and 47.6 percent, for comparison with the fuel cell and combined cycle cases. The lower performance of the plant utilizing Texaco MBG is due primarily to the lower MHD generator output resulting from the lower electrical conductivity associated with lower combustion temperature.

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TABLE 3.3.2-1

PERFORMANCE ESTIMATES FOR MHD SYSTEMS

	Texaco	BGC
Fuel System:		
Flow Rate (1b/sec)	390.4	282.7
Heating Value, HHV (Btu/lb)	5282	7290
MHD Generator:		
Oxygen Enrichment (% by Vol.)	35	35
Oxidizer Flow Rate (1b/sec)	652	693
Oxidizer Preheat Temp. (^O F)	1100	1100
Fuel Preheat Temp. (^O F)	1100	1100
Seed (% K by Wt.)	1.0	1.0
Pressure Ratio (Compressor)	11.5	10.5
Combustor Exit Temp. (^O F)	4650	4750
Magnet Strength (Tesla)	6	6
MHD Gen. Output, dc (MW)	510	560
Steam Cycle:		
Throttle/Reheat Temp. (^o F/ ^o F)	1000/1000	1000/1000
Throttle/Reheat Pres. (psi/psi)	2400/444	2400/444
Throttle/Reheat Flow (lb/sec/lb/sec)	913/689	886/668
Power Output (MW)	641	622
Overall Plant:		
Total Power (MW)	1151	1182
Net Power (MW)	944	969
Heat Rate, HHV - basis (Btu/kwh)	7864	7652
Heat Rate, LHV - basis (Btu/kwh)	7368	7164
Efficiency, HHV - basis (%)	43.4	44.6
Efficiency, LHV - basis (%)	46.3	47.6

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3.3.3 References

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Task 4.3 Evaluation of Integrated Gasification MHD Power Systems

The objective of this task is to determine the overall power plant efficiency of an integrated gasification MHD (IGMHD) power plant. Two types of gasifiers will be considered and the performance of the two resulting IGMHD power plants will be compared. Flow sheets for the gasifiers, the oxygen-blown Texaco and British Gas Corp. slagging types, are those previously developed by the contractor as agreed upon by NASA and the contractor.

4.3.1 Identification of IGMHD Power Plant with Texaco Gasifier

The MHD power system described in Section 3.3.2 utilizing medium-Btu gas (MBG) derived from the oxygen-blown Texaco gasifier will be used as the basis for the IGMHD power plant. The operating conditions are the same but the equipment output differs because of different steam loads in the power plant and power consumption by the gasification system. Figure 4.3.1-1 is a schematic for the interfacing between the gasification plant and the MHD power plant. Figures 4.3.1-2 and 4.3.1-3 are schematic diagrams of the MHD plant.

The fuel gas leaving the gasifier (1) is split, one stream being water quenched (2) to remove slag while the other passes through a high-pressure boiler (3). Prior to going through the boiler, the hot gas temperature is moderated by a recycle stream of cool gas (9). The steam raised in the high-pressure boiler (19) is superheated in the MHD boiler and then goes to the high-pressure turbine in the steam cycle of the MHD power plant. The fuel gas from the high-pressure boiler goes to a fuel gas regenerator (4), an intermediate-pressure boiler (5) and a low-pressure boiler (6).

The steam from the intermediate-pressure boiler (20) is combined with steam from the sulfur plant boiler. It is superheated by means of the exhaust gas from the MHD generator and goes to the intermediate pressure turbine of the MHD plant. Steam from the low pressure boiler is split into two streams. One stream is used for the gasification process and the other is fed to a feedwater heater (HTR.2) in the MHD steam bottoming cycle. This reduces the amount of extraction steam from the steam plant turbines and raises the turbine output.

The fuel gas is scrubbed (7) and parses through a fuel gas regenerator (9) to a feedwater heater (10), allowing further reduction in the extraction steam required from the MHD plant steam turbines. This feedwater heater allows elimination of HTR which had appeared in the Fig. 3.3.2-2. Additional heat exchangers are needed to reduce the gas temperature prior to sulfur removal.



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Following desulfurization, the low-temperature fuel gas passes through the two regenerators (14, 15) to raise the temperature of the fuel gas to 1100 F at the inlet to the fuel-gas expander (16). The expander lowers the fuel gas pressure to the level required for injection into the MHD combustor and provides additional shaft power. Since the expansion also reduces the temperature, additional heat is provided from the MHD exhaust flow (17, 18) to bring the fuel gas temperature up to 1100 F, which is required to produce a combustion temperature adequate to achieve high electrical conductivity of the plasma. The estimated performance for the Texaco-based IGMHD plant is given in Table 4.3.1-1.

4.3.2 - Identification of IGMHD Power Plant with BGC Gasifier

The MHD power system described in Section 3.3.2 utilizing medium-Btu gas (MBG) derived from the oxygen-blown BGC gasifier will be used as the basis for the IGMHD power plant. The operating conditions are the same but the equipment output differs because of interchanges of flows and energy between the gasifier and the power plant. Figure 4.3.2-1 is a schematic for the interfacing between the gasification plant and the MHD power plant. Figures 4.3.2-2 and 4.3.2-3 show the corresponding schematics for the MHD plant.

The fuel gas leaving the gasifier undergoes a quench (1) and transfers heat for feedwater heating (2) prior to sulfur removal (3). The heating of the MHD steam bottoming cycle feedwater reduces the amount of extraction steam from the steam plant turbines eliminating HTR.1 (shown in Fig. 3.3.2-2) and raising the turbine output. Heat is provided to the clean fuel gas (4) from the MHD \propto haust gas, raising the fuel gas to 1100 F at the inlet to the fuel gas expander (5). The expander lowers the fuel gas pressure to the level required for injection into the MHD combustor and provides additional shaft power. Since the fuel gas temperature drops as it goes through the expander, additional heat is provided from the MHD exhaust gas to raise the temperature back up to 1100 F. The estimated performance for the BGC-based IGMHD plant is given in Table 4.3.1-1.

4.3.3 Comparison of Performance

The performance estimates in Table 4.3.1 indicate that the IGMHD power plants using the Texaco and BGC gasifiers would be 34.4 and 37.2 percent, respectively, based upon the fuel higher heating value. The corresponding efficiencies based upon the fuel lower heating value would be 35.9 and 38.9 percent. The higher efficiency of the BGC-based case results from both the higher efficiency of the BGC gasifier and the higher MHD power plant efficiency when BGC medium-Btu gas is the fuel.

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TABLE 4.3.1-1

PERFORMANCE ESTIMATES FOR IGMHD SYSTEMS

	Texaco	BGC
Fuel System:		
Coal Rate (tons/day)	9414	7968
Heating Value, HHV (Btu/lb)	12,235	12,235
MHD Generator:		
Oxygen Enrichment (% by Vol.)	35	35
Oxidizer Flow Rate (lb/sec)	652	693
Oxidizer Preheat Temp. (^O F)	1100	1100
Fuel Preheat Temp. (⁰ F)	1100	1100
Seed (% K by Wt.)	1.0	1.0
Pressure Ratio (Compressor)	11.5	10.5
Combustor Exit Temp. (^O F)	4650	4750
Magnet Strength (Tesla)	6	6
MHD Gen. Output, dc (MW)	510	560
Steam Cycle:		
Throttle/Reheat Temp. (^o F/ ^o F)	1000/100	1000/1000
Throttle/Reheat Pres. (psi/psi)	2400/444	2400/444
Throttle/Reheat Flow (lb/sec/lb/sec)	1152 /9 92	886/648
Power Output (MW)	664	539
Overall Plant:		
Total Power (MW)	1174	1099
Net Power (MW)	967	886
Heat Rate, HHV - basis (Btu/kWh)	99 25	5174
Heat Rate, LHV - basis (Btu/kWh)	9498	8773
Efficiency, HHV - basis (%)	34.4	37.2
Efficiency, LHV - basis (%)	35 .9	38.9

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FIG. 4.3.2-1



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INTEGRATED BGC GASIFICATION/MHD POWERPLANT

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STEAM AND FEED WATER CYCLE CONFIGURATION FOR INTEGRATED BGC GASIFICATION/MHD POWER PLANT

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4.3.4 Load Following Capabilities of IGMHD Power Plants

The capability of an MHD power plant to respond to rapid load changes is determined by the response characteristics of the steam bottoming plant and gasifier plant since the MHD generator itself has very fast response characteristics. In fact, MHD generators have been proposed for applications which require rapid pulsed power output. On the basis of response-time capability, the IGMHD power plants are, therefore, estimated to be able to meet electric utility response time requirements.

An estimate of part load performance of MHD power plants indicates that at 70 percent of rated load, the efficiency of an MHD power plant to 96 percent of its full load efficiency. At 50 percent of rated load, the efficiency was estimated to decrease to 83 percent of its full load efficiency.

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A.1 Fuel Cell Electrochemical Fundamentals

Fuel cells are electrochemical energy conversion devices, defined as electrochemical cells that can directly and continuously transform the chemical energy of a fuel and oxidant to electrical energy by an isothermal process involving an essentially invariant electrode-electrolyte system. Unlike a battery, a fuel cell in theory does not require recharging rather it will operate as long as both fuel (H₂ from a hydrocarbon fuel) and an oxidant (O₂ from air) are supplied to the electrodes. The electrodes act as reaction sites where the electrochemical transformation of the fuel and oxidant occurs, producing d.c. power. This electrochemical transformation is isothermal; that is, the fuel cell directly uses the available free energy in the fuel at its operating temperature. Thus, it is not Carnot-cycle limited and can yield a high fuel-to-d.c. power conversion efficiency.

To comprehend the operation of a fuel cell, it is necessary to understand the basic electrochemical relationships which incorporate the parameters of primary importance in the determination of fuel cell performance. Because the primary parameters in these relationships do effect performance, an understanding of them will also provide an explanation for the thrust for the fuel cell development activity.

The following presents a brief introduction to the electrochemistry of fuel cells. This presentation has drawn upon the presentation of Ref. A.1-1 which provided a commentary at the appropriate level of detail for appreciating the discussion advanced in the current study. A more exhaustive treatment of electrochemical theory can be obtained from Refs. A.1-2 through A.1-9.

A.1.1 Reversible (Equilibrium) Electrode

Consider an oxidation-reduction reaction such as in Eq. A.1.1-1 occurring at an electrode:

 $aA + ne^- \neq bB$ (A.1.1-1)

At equilibrium, there is no net reaction, but there is a difference in potential between the electrode and electrolyte. Based on the following thermodynamic relationships:

$\Delta G = -nFE$	(A.1.1-2)
$\Delta G = \Delta G^{O} + RT \ln K$	(A.1.1-3)

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where: \(\Delta G = free energy change\)
n = number of electrons transferred in a reaction
F = Faraday constant
E = fuel cell equilibrium potential
\(\Delta G^{O_m}\) standard free energy change
R = universal gas constant
T = absolute temperature
\(K = equilibrium constant\)

Nernst developed an equation relating the equilibrium potential, E, to the activities of the species involved in the reaction. In general, the Nernst equation is written as:

$$E = E^{O} + \frac{RT}{nF} \ln \frac{[\text{product of the activity of reactant species}]}{[\text{product of the activity of product species}]}$$
(A.1.1-4)

where E^0 is the standard potential when all the species involved are at unit activity. The standard potential, E^0 , is related to the standard free energies of the species by:

$$\Delta G^{O} = -nFE^{O} \tag{A.1.1-5}$$

For the acid fuel cell, the overall reaction is given in the following equation:

$$H_2 + 1/2 0_2 \neq H_2 0$$
 (A.1.1-6)

At the normal operating temperature of 190 C and $E^0 = 1.14$ volts, the Nernst equation is: 1/2

$$E = 1.14 + \frac{RT}{2F} \ln \frac{\left[P_{H_2}\right] \left[P_{O_2}\right]^{1/2}}{\left[P_{H_2}\right]}$$
(A.1.1-7)

The molten carbonate reactions are given in the following equations:

 $H_2 + CO_3^{-7} \neq H_2 0 + CO_2 + 2e^{-7}$ (A.1.1-8)

 $CO_2 + 1/2 O_2 + 2e^- \neq CO_3^-$ (A.1.1-9)

Because CO₂ is fed to the cathode and produced at the anode, the concentrations are not necessarily equal, so the overall reaction is written as:

$$H_2 + [CO_2]_c + 1/2 O_2 + [CO_2]_a + H_2O$$
 (A.1.1-10)

At a normal operating temperature of 650 C and E^{O} = 1.02 volts, the Nernst equation can be written as:

$$E = 1.02 + \frac{RT}{2F} \ln \frac{[P_{H_2}]}{[P_{CO_2}]_a [P_{H_2O}]} + \frac{RT}{2F} \ln [P_{CO_2}]_c [P_{O_2}]^{1/2}$$
(A.1.1-11)

The fuel cell equilibrium potential, E, can be less or greater than the standard potential, E^{O} , depending on the sign and magnitude of the logarithmic, concentration-dependent terms of Eqs. A.1.1-7 and A.1.1-11.

Current is related to the electrochemical reaction rate by:

where r is the reaction rate, i/nF. At equilibrium, the forward and reverse reaction rates of Eq. A.1.1-1 are equal. They produce currents equal in magnitude but opposite in sign. The magnitude of the currents is called the equilibrium exchange current density, i_0 . High exchange current density indicates that the electrochemical reaction rate is high and good fuel cell performance is possible.

A.1.2 Non-Equilibrium Electrodes - Polarization

Chemical and physical factors associated with various elements of the cell when current is flowing limit the processes described above. This results in a reduction in voltage from the expected reversible potential, E, (Fig. A.1.2-1). Knowledge of these limitations, called polarizations, overpotentials, or overvoltage 3, n, is useful when attempting to calculate efficiencies.

$$\eta = E - V$$
 (A.1.2-1)

where V is the cell voltage.

The total polarization of the electrode is the sum of the factors which effect efficiency.

A.1.2.1 Activation Polarization - Tafel Equation

Most chemical reactions involve an energy barrier that must be overcome by the reacting species. This activation energy results in activation polarization, n_{act} . Activation polarization is the energy loss which occurs as the fuel or oxidizer is transformed into a state suitable for reaction. This energy loss is the sum of the heat of chemisorption on the catalyst and the heat of formation of the active ion. It is related to the current, i, by the Tafel equation:

$$\eta_{act} = a + b' \log i$$
 (A.1.2.1-1)

This can also be written as:

$$i = i_0 \exp(2.3\eta_{oot}/b')$$
 (A.1.2.1-2)

where a and b' are empirical and b' is referred to as the Tafel slope obtained from plots of log i versus n. The exchange current can be determined experimentally by extrapolating plots of log i versus n to n = 0. Equation A.1.2.1-2 describes either the fuel cell anode or the cathode when the appropriate values for i_0 , n, and b' are used.

Activation polarization is related to the slow step in the reaction sequence. The slow step could be adsorption of reactant onto the surface, electron transfer, desorption of product, or any other step in the reaction sequence.

Since the required reaction rate for chemisorption and ion formation are both proportional to the current density and a greater potential is needed to maintain a high reaction, the activation polarization loss increases with current density. Porous electrodes of large internal surface area are used to increase the number of "active sites" for chemisorption and, thus, to decrease the potential required for a given over-all reaction rate. An other approach to decreasing activation polarization is to increase temperature.

A.1.2.2 Concentration Polarization - The Limiting Current

Concentration polarization occurs when the electrode reaction is hindered by mass transport effects. The controlling process in the case of concentration polarization may be slow diffusion in the gas phase through the pores of the electrode, solution/dissolution of reactants/products into/out of the electrolyte, or diffusion through the electrolyte to a reaction site. As practical current densities, the latter predominates. When the electrode is completely governed by diffusion processes because of low reactant concentration

in the feed gases or because of conversions approaching 100 percent (reactants depleted), the limiting current, i_L is reached. This situation is characterized by a rapid drop in cell voltage, with increasing current density, as shown in Fig. A.1.2.2-1.

The limiting current, i_L , can be calculated from the diffusion coefficient of the reacting ions, D, the bulk concentration of the reacting ions, C_B, and the thickness of the diffusion layer, X, by applying Fick's Law:

$$i_{\rm L} = \frac{{\rm DnFC}_{\rm B}}{{\rm X}}$$
 (A.1.2.2-1)

For an electrode free from activation polarization (that is, the reaction is fast), the concentration overpotential is X:

$$\eta_{\rm conc} = \frac{RT}{nF} \ln \left(1 - \frac{1}{1L}\right)$$
 (A.1.2.2-2)

It should be noted that it is possible to properly design a fuel cell so that the concentration polarization effect on cell voltage is negligible.

A.1.2.3 Total Electrode Polarization

The total overpotential, η , is the sum of the activation and concentration polarizations.

$$\eta = \eta_{act} + \eta_{act}$$
 (A.1.2.3-1)

In Eq. A.1.2.3-1, E_e is not the Nernst potential for a complete cell (Eqs. A.1.1-7 or A.1.1-11), but is the Nernst potential of either the oxidation or reduction half-cell reaction, as measured against an inert reference electrode.

The signs of the quantities in Eq. A.1.2.3-1 are always such that $V_e \leq E_e$ so that the equation is applicable to both anode and cathode.

A.1.2.4 Ohmic Polarization - iR Losses

When an anode and cathode are connected to an external load, the circuit is complete and current flows. Resistance to conduction of ions through the electrolyte, electrons through the electrodes and current collectors, and by contact resistances (between components) causes ohmic or iR polarization. Because both fuel cell electrodes and the electrol/te obey Ohm's Law, it is easy to calculate the voltage gradient (and loss) required to force ionic and electronic conduction: and side of



FUEL CELL POLARIZATION CURVE

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(A.1.2.4-1)

where R is the total cell resistance including electronic, ionic, and contact resistances.

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A.1.3 Cell Performance

n_{ohm} = iR

Both anodic and cathodic reactions contribute to the cell voltage, and the internal resistance of the cell reduces the available voltage. Therefore,

 $V_{cell} = V_{anode} + V_{cathode} - iR$ (A.1.3-1)

where Vanode and V cathode are determined experimentally or by Eq. I.A.2-7.

Polarization cannot be eliminated, but can be minimized by cell design. The iR loss can be reduced by using thin electrolytes and be efficient contacts between electrode and leads. Temperature and electrolyte composition also influence the cell resistance.

Concentration polarization is dependent on the mass transport properties of the system. Mass transfer is a function of temperature, pressure, concentration, and the physical properties of the system. In a fuel cell, the reactants must diffuse through the tortuous paths of the porous electrode. This makes electrode structure important. The electrochemical reaction rate is a function of the concentration of the reacting species so, as the conversion increases, the concentration polarization becomes more severe.

Activation polarization depends upon the fundamental electrochemical reaction rate, which is also a function of temperature, pressure, and concentration. However, because the fuel call reactions under consideration are catalytic, the activation polarization can be changed by using different catalysts.

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B.1 Potential Fuel Cell Power Generation System Advantages

Electric power plants containing a fuel cell power generation system would have many characteristics which could be attractive to electric utilities. Among the more important characteristics are:

- 1) high fuel efficiency at both design rating and over the plant's operating range,
- 2) low environmental emissions,
- 3) rapid response to load-demand changes,
- 4) waste heat recovery potential, and
- 5) little effect of size on efficiency.

The following commentary elaborates on the potential benefits which could occur to an electric utility from these characteristics.

B.1.1 Fuel Cell Efficiency

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Because a fuel cell is not a thermal machine, it is not limited by the Carnot cycle. It therefore offers the potential for higher conversion efficiencies and greater fuel conversion than conventional thermal generators are likely to achieve. First-generation fuel cells, (phosphoric acid) would have efficiencies of 38-40 percent, second-generation (molten carbonate and advanced phosphorus acid), 45 percent. In contrast, the efficiency of conventional fossil-fueled generators is in the range of 28-38 percent. As cell technology advances, these efficiencies are bound to increase. Also, since fuel cell efficiency is related to individual cell performance rather than generator size, small power plants will be just as efficient as larger ones (Ref. B.1-1).

Besides being more efficient than conventional generators, fuel cell power plants maintain their efficiency over a wide range of loads. This characteristic is displayed in Fig. B.1.1-1 where part-load heat rate is plotted for seven power plants including two fuel cell power plants. Whereas conventional power generation equipment is more efficient at rated power and markedly less efficient at part power, the fuel cell power plant is nearly constant in efficiency from 25 to 100 percent of its rated power output.

The high part-load efficiency could permit the fuel cell plant to be used in the load-following mode (Ref. B.1-2). By operating the fuel cell power plant in a load-following mode to provide for the variation in power demand a utility need not vary the output from their conventional intermediate or peaking units. This will then permit them to be operated at their most efficient output, 100 percen⁻ of rated power.



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COMPARISON OF POWER SYSTEM EFFICIENCIES

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B.1.2 Environmental Emissions

The fuel cell does not operate on a combustion cycle, consequently emission levels other than carbon dioxide, air, and water, such as nitrogen oxides (NO_x), carbon monoxide (CO), and unburned hydrocarbons (UHC), are not high. The emission of sulfur oxides (SO_x) are also not high because the sulfur content of the fuel stream must be reduced to attain long fuel cell life. The NO_x , CO, UHC, and SO_x emissions that do occur originate mainly in the power plant's fuel processor (kef. B.1-3).

The environmental emissions of fuel cell power plants are expected to be low. Table B.1.2-1 compares Federal Environmental Protection Agency (EPA) requirements with the design goals for the 4.8-MW unit being installed at Con Edison in N.Y. and the measured emissions for a 1-MW test unit using naphtha as fuel. It can be concluded from this table the emission from a fuel cell power plant are expected to be at least an order of magnitude below EPA standards.

In addition, the fuel cell is expected to be quiet. There will be pumps and cooling fans, but the noise level will be appreciably below the level of a fossil steam plant (Ref. B.1-4). Also little or no makeup water or cooling water will be required eliminating the need for cooling ponds.

B.1.3 Response Characteristics

Fuel cells are quick to follow load changes. The load response time for the 4.8-MW demonstration to be installed in the Con Edison system has a performance specification of minimum power to rated power of 15 seconds and a specification of 35-percent power to rated power of one second (Ref. B.1-3).

If a fuel cell were operated in the load-following mode to take advantage of its quick response, it could be used to reduce mechanical spinning reserve requirements. Mechanical spinning reserve is supplied by a conventional power plant operating in a part-load condition so that it has the capability to match load variations. Operation in this mode is, however, is relatively inefficient due to the poor off-design characteristics of most conventional power plants. Reducing the mechanical spinning reserve by substituting the fuel cell with its rapid response and superior part-load performance characteristics would result in utility production cost savings.

B.1.4 Waste Heat Recovery Potential

Fuel cells because of their elevated operating temperature produce waste heat that can be utilized in the production of hot water or steam. The hot water could be used in a cogeneration application to supply district heat or

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TABLE B.1.2-1

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FUEL CELL POWER PLANT EMISSIONS POUNDS PER MILLION BTU ENERGY INPUT (from Ref. B.1-3)

Emission	EPA Requirement	4.8-MW Design Goal*	l-MW Test Results
NO.	0.3	0.02	0.02
so	0.8	0.00003	0.0015
Particulates	-	0.00003	Non-Detectable
Smoke	-	None	None

* At rated power

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industrial process heat while the steam could be used in a steam bottoming cycle power plant to produce additional electric power. Utilization of this waste in a cogeneration application could increase overall efficiency to as high as 90 percent (Ref. B.1-1) while the bottoming cycle application of the waste heat could yield overall efficiencies between 50 percent (phosphoric acid) and 75 percent (molten carbonate).

B.1.5 Scaling Effects

The heart of a fuel cell power generation system, the fuel cell stack, consists of a stack of individual fuel cells connected in series to obtain the desired voltage. Since all the individual cells operate essentially identically, the efficiency of the cell stack is comparable to the individual cell. Consequently, for the voltages and the cell stack outputs envisioned for electric utility applications the fuel cell stack efficiency is independent of output.

The lack of a scale effect has numerous advantages. It permits the development of a standard size module in the 5-25 MW range which could be factory-fabricated and mass produced. This would reduce costs and lag time between orders and commissioning to about 24 months (Ref. B.1-1). The standard size modules would allow the utility to tailor their additions to more closely match anticipated load increases. Once installed the presence of many smaller modules would increase system reliability in comparison to one large power plant with the name output. This, in turn, reduces the magnitude of reserve requirement needed to cover for forced outages.

Additional benefits accrue due to the coupling of modularity and low environmental emissions. The tailored power plant can be located near the load center thus reducing transmission losses and costs. The reduced transmission distances will permit easier control of reactive power and voltage. Lastly, easier siting increases the probability of effective waste heat utilization. ł

B.1.6 References

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C.1 Fuel Cell Historical Overview

The fuel cell concept is an old concept. In 1839, W. Grove, building upon earlier work performed by H. Davy, constructed a chemical battery in which the waterforming reaction of hydrogen and oxygen generated an electric current. Grove used platinum electrodes in contact with dilute sulfuric acid in his device. Fifty years later, L. Mond and C. Langer developed a device they called a fuel cell which had an output of 6 amps/ft². The development of the dynamo in the same time period overshadowed this work and the fuel cell remained undeveloped for another fifty years. In the early 1940's Justi started to work with porous metal electrodes. Finally, in 1954, F. Bacon published the results of his high-pressure cell (Ref. C.1-1).

The real impetus for fuel cell development occurred as a result of the NASA-sponsored programs for the Gemini and Apollo spacecraft in the early 1960's. By the mid-1960's some 50 U.S. companies were investigating fuel cells as these companies tried to adapt the fuel cell to non-space applications. By the late 1960's the number of participants had dwindled because most companies were unwilling to make the long-term research and development commitments required to develop the fuel cell for commercial applications (Ref. C.1-2).

One of the companies willing to make the commitment has been United Technologies Corporation (UTC, formerly United Aircraft). Although it may appear somewhat provincial in outlook, it can be stated that this commitment has placed UTC in the forefront of fuel cell activity. It can be further stated that a review of UTC fuel cell activities would essentially cover most of the highlights associated with fuel cell development in the last quarter century.

C.1.1 Early UTC Activity

Energy systems studied at United Technologies in the 1950's indicated that fuel cells would be attractive as a source of on-board power for space vehicles and that they also had potential for many military and commercial applications. A patent license was obtained for the Bacon type of fuel cell which was the most advanced at the time and which was applicable to NASA missions. This patent license took the form of an agreement with National Research and Development Corporation in Britain who was the owner of the Bacon patents and the Patterson-Moos Laboratories (now Leesona Corporation) who held a United States license. Certain obligations of this agreement survive to the present.

C.1.2 NASA Program History

NASA gave early support to the fuel cell program and has continued to fund the application of fuel cells to space craft through the Apollo Program, Space Shuttle Procurements, and many R&D programs.
In March 1962, North American Aviation, prime contractor for the NASA Apollo spacecraft, contracted with Pratt & Whitney Aircraft to develop and produce 1.5 kilowatt fuel cell power plants for lunar missions. After successful development, 61 power plants were manufactured and delivered through 1968. Over 10,000 failure-free hours were experienced in space including all Apollo Lunar, Apollo Sky-Lab, and Apollo-Soyuz missions.

In November 1973, Pratt & Whitney Aircraft was named to develop and produce fuel cell power plants for the NASA Space Shuttle Orbiter under subcontract with Rockwell International Corporation (formerly North American Aviation). The Space Shuttle (rbiter fuel cells have been constructed and have demonstrated the required performance for this contract. Drawing on technology developed since 1962, the Orbiter develops 12 kilowatts of power compared to 1.5 kilowatts for Apollo and has a weight of 201 pounds compared to the Apollo all-up weight of 248 pounds. The Shuttle power plant will have a required lifetime of 2,000 hours (5,000 hours have been demonstrated) compared to the useful life of 450 hours required of the Apollo power plant (Ref. C.1-3).

C.1.3 Other Space and Defense Programs

A contract was received from NASA-Marshall Space Flight Center to deliver a power plant similar to the Shuttle Orbiter fuel cell for Space Tug, an auxiliary spacecraft in the Shuttle Program. NASA-Lewis Research Center has continued to support advanced space fuel cell technology contracts to develop ultra-lighweight fuel cells.

A power plant system including two fuel cell power plants and deep submergence-type fuel tanks was completed and delivered to the Navy for the Deep Submergence Rescue Vehicle. A submersible auxiliary power unit for a classified mission based on this DSRV design was delivered in August 1975. Previous programs with the U.S. Army for portable power supply and with the Air Force for airborne weapons power supplies have lapsed largely because of lack of mission definition in those agencies.

C.1.4 Fuel Cell Program for the Gas Utilities

Columbia Gas became aware of United's interest in fuel cells through the gas turbine pipeline pumping program. They undertook a three year program in 1963 to investigate natural gas fuel cells. Following the test of a 4-kW experimental fuel cell power plant by Columbia Gas Systems in 1966, some thirty gas utilities formed a not-for-profit corporation called TARGET (Team to Advance Gas Energy Transformation) to support fuel cell research at UTC. During 1971-73, 65 experimental 12.5-kW fuel cell power plants (PC-11 designation), designed and built by UTC in conjunction with TARGET, were

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installed and operated by gas and combination utilities in 35 on-site locations to determine the conceptual feasibility of on-site fuel cell electric service. The longest continuous run recorded without major overhaul in these tests was 4,000 hours; 10,560 hours was the longest run on any single power plant with overhaul; and a total of 200,000 hours was logged in the program. This testing, conducted by gas industry personnel at gas industry expense, provided service experience with the real impurities of natural gas and water supplies, furnished actual market acceptance and public relations experience, and demonstrated that gas personnel can be trained to install and operate fuel cell type equipment. Many mechanical problems were experienced early in the program but by the end of the program a failure rate of less than one unscheduled shutdown per thousand hours was experienced, which is comparable to industrial gas turbine experience. This experimental power plant identified a number of system requirements which provided the objectives and direction for the continuing effort.

In 1976, a 40-kW experimental unit (PC-18) reflecting lessons learned from the previous comprehensive field tests, successfully demonstrated electrical generating efficiency of 40-percent and the availability of power plant waste heat at useful temperatures. The combined output of electrical and thermal energy exceeded 80 percent overall natural gas utilization in contrast to about 30-percent utilization for electrical power plants and 60-percent utilization for space and water heating.

An even higher utilization level was achieved in testing involving a simulated 16-unit apartment building located in New England. Winter conditions were simulated and a heat pump was included with the heat recovery system. The results produced indicated that less gas energy would be used to provide the total energy requirements of the building -- electrical and thermal -- than was needed by a conventional gas-fired furnace to provide heat alone.

The successful 18,000 hours of operation of this power plant in United's facility and continuing gas utility and UTC expenditures and advocacy resulted in sponsorship by the Gas Research Institute (GRI, the successor to TARGET) and DOE of a 40-kW development program which was initiated in 1977. This program incorporated recent technological improvements in the development of an improved 40-kW design suitable for precommercial field test.

In the near future, field tests of some fifty 40-kW fuel cell systems will be conducted at some 25 sites by a similar number of utilities. In parallel with these tests, it is proposed that the utilities examine proposed business scenarios for commercial service, and the Federal Government and GRI determine what incentives are required, if any, to encourage commercialization. These actions would provide the basis for establishing the detailed plans and actions required for commercialization (Ref. C.1-4).

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C.1.5 Fuel Cell Program for the Electric Utilities

In 1971 the Edison Electric Institute (EEI), and a group of electric utilities began evaluating the fuel cell. Significant benefits were identified, and in 1972 the Fuel Cell Generator-1 (FCG-1) program was mounted by United and nine utilities. The goal of the program was the development of a 26-MW, power plant for commercial service by 1980. Operational targets included a capital cost of \$350/kW in 1980 dollars (the stated goal was \$185/kW in 1972 dollars), a heat rate of 900C-9300 Btu, a 40,000-hour stack life, and the ability to run on naphtha (Ref. C.1-2).

A 1-HW pilot FCG-1 plant was demonstrated successfully in 1976-1977 for more than 1,000 hours generating about 700,000 kWh (Ref. C.1-5). The plant, while not in the packaged, modular form these plants will ultilately take, confirmed that a mephtha-fueled unit could provide electricity to a utility bus, while meeting heat rate, loading-following, emissions, and other operational requirements established by the utilities.

In 1976 BRDA (now DOE) and the Electric Power Research Institute (EPRI) joined the FCG-1 effort, resulting in a program to design, fabricate, and test one 4.5-MW a.c. module of the FCG-1 power plant in 1979-1980. Consolidated Edison Company of New York, was chosen as the host utility, and a downtown site in New York City was selected for the demonstration. The 4.5-MW demonstrator is expected to exhibit the suitability of the fuel cell for utility service. Unlike the 1-MW plant, the 4.5-MW unit will be fully packaged, with virtually all components organized into modules. After fuel cell stack installation in 1981 an acceptance test of 200 hours will commence. This will be followed by a 2000-hour validation test (Ref. C.1-5). Plugged into Consolidated Edison's grid, the demonstrator will be operated by utility personnel. The plant's downtown location is expected to verify the fuel cell's emissions, and noise levels. In addition, general acceptability to its neighbors will be evaluated. Economic advantages in the form of transmission savings, voltage control, load following, and system reliability will be determined. The program will also verify the FCG-1's operational claims, including those on heat rate, power quality, transient response, and startup-shutdown characteristics (Ref. C.1-2).

C.1.6 Summary

A chronological summary of the fuel cell activity mentioned above is provided in Table C.1.6-1.

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TABLE C.1.6-1

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CHRONOLOGY OF UTC FUEL CELL ACTIVITIES

- 1959 Bacon Patent Licence Obtained - U. S. Army Regenerative Cell Contract
- 1961 NASA 250-W Fuel Cell Contract Columbia Gas 500-W Cell Contract
- 1962 Apollo Development Contract
- 1963 Columbia Gas Hydrocarbon-Air Cell Contract
- 1966 3.75-kW Natural Gas Powerplant Delivered to Columbia Gas - First Apollo Fuel Cell Flight
- 1967 TARGET Gas Utility Program Initiated
- 1969 TARGET 12.5-kW Powerplant Tested - First Apollo Lunar Mission
- 1970 Navy Deep Submergence Contract
- 1972 Electric Utility Development Initiated
- 1974 Space Shuttle Orbiter Contract
- 1975 TARGET 40-kW Powerplant Tested
- 1976 ERDA (DOE) and EPRI Contract
- 1977 DOE and GRI Contract - FCG-1 Pilot Power Plant Tested
- 1981 DOE/EPRI 4.8-MW Demonstrator Installed - Possible DOE/GRI 40-kW Field Tests

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A.2 Theory of Combined-Cycle Operations

In the simplified integrated gatification combined-cycle system, the turbine exhaust gas is used directly to raise steam in an unfired boiler. Additional heat from the fuel gas stream is also used to raise steam. The performance of such a system can be estimated from the following equation:

$$\eta_{cc} = \frac{\eta_{gc} + \left[\left(1 - \eta_{gc}\right) + \frac{W_g}{W_f} \right] \left(\frac{T_{ex} - T_{et}}{T_{ex} - T_e} \right) \eta_g}{1 + \frac{W_g}{W_f}}$$
(A.2-1)

where

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In Equation A.2-1, the ratio W_g/W_f represents the amount of steam raised by heat other than that contained in the gas turbine exhaust. This heat may be the fuel gas sensible heat used to raise steam or, in some configurations, fuel gis may be burned in the turbine exhaust prior to the waste heat boiler to give a higher approach temperature.

For the case where W_g is zero, i.e., the pipeline delivery of fuel gas, the equation A.2-1 simplifies to:

$$\Pi_{cc} = \Pi_{gt} + \left[(1 - \Pi_{gt}) \left(\frac{T_{ex} - T_{gt}}{T_{ex} - T_{gt}} \right) \right] \Pi_{g} \qquad (A.2-2)$$

These equations point out the importance of gas turbine efficiency and of cycle temperatures on the ability to utilize the gas turbine exhaust heat.

The gas turbine efficiency is a function of pressure ratio, turbine inlet temperature and the mechanical efficiencies of the components. Losses due to internal pressure drops, cooling loads, diffuser losses, etc. must also be accounted it 1

For the gas turbine used in this study, the efficiency is in the 32 to 34 percent range.

Similarly, the steam turbine cycle efficiency is a function of operating pressure, throttle temperature, cycle configuration (reheat, regeneration and component efficiencies). A fully regenerative single reheat steam cycle operating at modern steam conditions (1800 - 2400 psi, 1000 F/1000 F) would have cycle efficiencies of 40-43 percent. In combined-cycles, however, there is usually no regeneration since gas turbine exhaust heat is used to preheat feed water, deaerate, etc. Steam cycle efficiencies are then of the order of 30-35 percent. The loss of cycle efficiency is more than made up by the reduction in stack temperature of the gas turbine exhaust.

All of the above parameters are very closely interrelated. It is virtually impossible to change one variable without affecting another. For this reason, a very sophisticated computational program must be used to reduce analysis times to reasonable levels. This is especially true when considering the integrated power plant because the operation of the gasifier must be accounted for.