

GULF CANADA RESOURCES INC.

MOUNT KLAPPAN COAL PROJECT

1982

POWER STUDY

Coal Licence Number 7118 to 7177

7381 to 7392

and

7416 to 7432 inclusive

Cassiar Land District

NTS Map Number 104 H

Latitude Between 57º11' and 57º22'N Longitude Between 128°39' and 129°05'W

BY

PHILLIPS BARRATT KAISER Engineering Ltd. Vancouver, B.C. Canada

PBK Project No. 82062

DECEMBER 1982



MOUNT KLAPPAN COAL PROJECT





 \smile







TABLE OF CONTENTS

INTRODUCTION

SUMMARY

PART

- 1 EXISTING AND POTENTIAL LOADS
- 2 COAL-FIRED ELECTRIC GENERATING STATIONS CAPITAL COST ESTIMATES
- 3 COAL-FIRED ELECTRIC GENERATING STATIONS OPERATING COST ESTIMATES
- 4 HIGH VOLTAGE TRANSMISSION SYSTEMS
- 5 DIESEL-POWERED ELECTRIC GENERATING STATIONS
- 6 ENGINEERING AND CONSTRUCTION SCHEDULES

APPENDIX A

LOCATION MAP OF MINING PROPERTIES

INTRODUCTION

In November, 1982, Gulf Canada Resources Inc. commissioned Phillips Barratt Kaiser Engineering Ltd. to prepare a study of the size and costs of a coal-fired electric generating plant which would be located at a coal mine at Mount Klappan. The powerplant would provide power for the coal mine and power for sale to existing and potential mines and towns in the region.

The coal mine is the subject of a separate Mine Assessment Study prepared for the Mount Klappan Coal Project.

Capital and operating cost estimates, of order-of-magnitude accuracy, and design and construction schedules have been prepared for the various sizes of powerplant necessary to satisfy the power requirements identified, together with the high-voltage transmission systems to serve the local mines and towns. In addition, in order to provide comparative costs to prospective mines of generating their own power, estimates of capital and operating costs of diesel-generator plants have been prepared.

The information developed in this study contains the basic elements of unit costs and potential power requirements of mines and towns selected. It is emphasised that the size of powerplant necessary depends upon the assumptions made as to which mines may be developed and the relative start dates for same. Similarly, the size of boiler-turbine units which should be installed initially to correspond with the rate of load growth can vary with the assumptions made. To provide a means by which cost estimates can be made of powerplants of varying installed capacities and sizes of units, capital cost curves and adjustment factors have been prepared, and examples are given to illustrate their use.

Cost estimates have been prepared for two alternative powerplant development schemes which correspond to the following mine production levels:

- Level A: 1 000 000 tonne per year. Coal supplied to the powerplant would be run-of-mine.
- Level B: 5 000 000 tonne per year. Coal supplied to the powerplant would be washed product.

Power demands from Mount Klappan mine activity have been obtained from the separate Mine Assessment study.

Recommendations have been included regarding further areas of study which could affect significantly the assumptions made and cost estimates prepared in this study.

At this stage of preliminary evaluation, standard utility technology has been used as the basis for cost estimating. Later studies would investigate the most appropriate systems.

SUMMARY

- 1. An initial review of the power requirements and location of existing and potential mines and towns lead to a selection of ten major B.C. mines for study as potential load centres for servicing with electrical power from a powerplant at Mount Klappan. There are two major areas of concentration of load:
 - Kutcho Creek Area, 125 kilometers north of Mount Klappan, involving four of the potential mines on the transmission route;
 - Schaft Creek Area, 150 kilometers west of Mount Klappan, involving three of the potential mines on the transmission route.

The four other mines are within a 400 kilometer radius.

2. Based upon an optimistic assumption as to anticipated start-up dates and sequence of development within the period 1987 through 2000, following are estimates of power demands of Mount Klappan coal mine operations and selected mineral mines:

Level A, 1 million tonne per year of run-of-mine coal shipped as product:

	M	ount Klappan MW	Mines <u>MW</u>	Total <u>MW</u>
-	Kutcho Creek area	13	50	63
-	Schaft Creek area	16	149	165
-	All ten mines	18-20	172-220	190-240

Level B, 5 million tonne per year of washed product coal:

		ount Klappan MW	Mines MW	Total MW
-	Kutcho Creek area	65	50	115
-	Schaft Creek area	71	149	220
-	All ten mines	72-74	173-221	245-295

- 3. Not included in the study are existing and potential mines in the Stewart/Kitsault area, the Yukon Territory and Alaska.
- 4. Information from various sources on reserves and production rates for potential mines as it related to power requirements was inconsistent. For this study, demand figures and start dates which had been provided to British Columbia Hydro and Power Authority (BCHPA) were adopted.

5. Capital cost estimates have been prepared for powerplants (constructed to utility standards) using coal with a 33% ash content (Level A), and with a 20% ash content (Level B). The costs per kilowatt of installed capacity are as follows:

Plant Size:	5	<u>0 MW</u>	100 MW	150 MW	200 MW	250 MW
Level A:	\$	2,800	2,250	2,120	2,000	1,900
Level B:	\$	-	2,100	1,970	1,860	1,800

All costs are in 1982 Canadian dollars.

- 6. The total cost of a powerplant necessary to provide a given firm capacity can vary with the sizes of units adopted to suit the load buildup schedule and standby capacity. Curves and factors have been provided to calculate these estimated costs to suit various combinations of unit sizes and standby capacity. The same curves may be used to estimate unit costs using coals varying in ash content between 20% and 33%.
- 7. The operating cost estimates in cents per kilowatt hour of the powerplants studied, and the corresponding unit costs of coal assumed, are as follows:

Plant Size: 5	<u>0 MW</u>	100 MW	150 MW	200 MW	250 MW
Level A		o 1 -			
\$25.00/tonne	2.44	2.17	2.02	1.99	1.93
\$50.00/tonne	3.95	3.66	3.45	3.46	3.39
Level B					
\$25.00/tonne	-	1.82	1.67	1.64	1.59
\$50.00/tonne	- ,	3.03	2.80	2.82	2.75
\$75.00/tonne	-	4.22	3.93	3.94	3.91

- 8. The capital and operating cost estimates of the high voltage transmission line systems were based upon discussions with BCHPA, who have performed route studies and cost studies for transmission systems from Mount Klappan to potential mine sites. Unit costs of 138 kV and 230 kV lines using wooden-pole H-frame construction, were estimated at \$140,000/km and \$160,000/km respectively. All systems are based upon a single line serving the selected mines, and no grid has been investigated to increase reliability of supply. The total cost of the transmission system required to serve the ten mines is \$245 million.
- 9. Capital costs of diesel generating plants of 10 MW, 25 MW and 50 MW installed capacity are estimated at \$820/kW, \$775/kW and \$755/kW respectively. Operating costs are estimated at 13.7¢/kWh, 13.4¢/kWh and 13.0¢/kWh respectively. Diesel fuel represents approximately 77% of the operating cost, assuming use of light diesel fuels.

- 10. Engineering and construction schedules, assuming that governmental approvals have been given, are as follows:
 - coal-fired powerplants could be designed and constructed in four to six years depending upon the size of plant and boiler delivery times;
 - for transmission lines, time required for land acquisition, and design and construction could be from three to four years;
 - diesel plants could take about two to two and one-half years to design and construct, depending upon the size of station.
- 11. In order to refine some of the broad assumptions made for this study and to supplement the findings, the following recommendations for further study are made:
 - identify potential site locations for a powerplant at Mount Klappan and establish their influence on costs;
 - identify potential sources of raw water and relative cost and reliability of supply;
 - identify potential cooling pond locations and construction, and compare costs of pond cooling with various tower cooling alternatives, including environmental considerations;
 - study methods of ash disposal specifically related to the site, incorporating environmental considerations;
 - in particular, prepare a study of environmetal and operational advantages of fluidized bed combustion boilers as applied specifically to Mount Klappan, and prepare cost estimates for same;
 - identify locations and costs of mini-hydro sites, presently being studied by BCHPA and others, as alternatives for the generation of power for potential major mines.
 - investigate the opportunities and advantages of district heating.

PART 1

()

 \bigcirc

()

EXISTING AND POTENTIAL LOADS

PART 1 - EXISTING AND POTENTIAL LOADS

T

1

CONTENTS

()

 \bigcirc

()

		Page
1.1	INTRODUCTION	1-1
1.2	INFORMATION SOURCES	1-1
1.3	LOAD ESTIMATES	1-2
	131 Mines	1-2
	132 Towns	1-2
	.1 Existing Towns	1-2
	.2 Potential Towns	1-2
	133 Mount Klappan Operations	1-4
1.4	LOAD GROWTH	1-5
	141 Introduction	1-5
	142 Load Growth Curves	1-5
	.1 Kutcho Creek Area	1-5
	.2 Schaft Creek Area	1-5
	.3 All Areas	1-6
1.5	YUKON AND ALASKA	1-6
1.6	OTHER POTENTIAL MINES	1-6
	161 Windy-Craggy	1-7
	162 Stewart/Kitsault Area	1-7
	163 Miscellaneous Properties	1-7
	.1 New Jersey Zinc Exploration	1-7
	.2 Gnat Pass	1-7
1.7	PRECIOUS METAL MINES	1-8

Table

- 1.1 Electrical Demand From Major Mines
- 1.2 Electrical Demand From Mining Towns
- 1.3 Mount Klappan Operations, Power Requirements

Figure

1.1 Load Growth Curves

Plate

1.1 Mine & Town Locations

1.1 INTRODUCTION

Major existing and potential mines have been identified for which it could be practical to have electric power provided from a coal-fired powerplant at Mount Klappan. A map indicating the location of the mines and Mount Klappan is provided in Plate 1.1.

For each mine, a potential electrical demand value has been assigned and load growth curves have been developed for various groupings of major mines. The load growth curves have been based upon an optimistic outlook for start-up of mines between 1987 and the year 2000. These curves include an allowance for power required for existing towns and settlements.

The power requirements to produce coal at Mount Klappan for sale have been obtained from the separate mine assessment study. The power requirements to produce coal to fuel the powerplants are developed herein.

1.2 INFORMATION SOURCES

Information on the mines was sought from industry and government sources by telephone and in meetings, and from published reports and other data. Information varied and was often qualified.

The anticipated life of each of the major potential mines was based on Table 3.3 of the British Columbia Regional Economic Study 1982, "The Northwest Region", produced by the Ministry of Industry and Small Business Development of B.C., May 1982. The demands projected for each mine identified were obtained informally, and are believed to correspond to those used by the Provincial Government in their studies.

A location map showing existing mines and mining properties in British Columbia and the Yukon Territory (including part of the Northwest Territories) is included in Appendix A. This map has been produced by the British Columbia and Yukon Chamber of Mines and is Revised January 1983.



1.3 LOAD ESTIMATES

131 MINES

The projected loads for the mines identified have been presented in Table 1.1. Also presented is an optimistic start date and expected life for each.

132 TOWNS

A listing of existing towns and settlements and potential towns is given in Table 1.2, together with estimated demands. (It may not be economical to service some existing towns: See Part 4.)

.1 Existing Towns

The towns in the region are quite small and, except for Cassiar, commercial and industrial loads are minimal. B.C. Hydro operates four diesel plants in the region as follows:

Town	Capacity (MW)
Atlin	1.75
Dease Lake	1.65
Eddontenajon (Iskut)	0.80
Telegraph Creek	1.15
Total	5.35

All four plants are multiple diesel-generator installations, some sets of which are mobile units. The loads served generally include only residential and small commercial customers. Typical corresponding reported populations served are:

Dease Lake - 130 families approximately Eddontenajon - 300 persons Telegraph Creek - 300 persons

The town of Cassiar is served by diesel generators owned by Brinco Ltd. (Cassiar Asbestos), with an installed capacity of about 12 MW. Brinco advised that the town has an approximate demand of 3 MW (with the mine requiring around 7 MW).

.2 Potential Towns

The estimated work force and electric demand at the mines are shown in Table 1.2. A load of 4.5 kW per worker has been assumed as a measure of peak demand from the towns.

TABLE 1.1

ELECTRICAL DEMAND FROM MAJOR MINES

(

()

 \bigcirc

Letain

Sustut

Logtung

Red Chris

Schaft Creek

Stikine Copper

Mine	Owner	Minerals	Optimistic Start Date	Expected Life-Years	Demand <u>MW</u>
Potential					
Adanac	Placer Development	Mo, W	1989	20	12
Eaglehead	Esso-Nuspar	Cu, Mo, Ag, Au	1995	15	13
Kutcho Creek	Esso-Sumitomo	Cu, Ag, Zn	1987	12	13
Letain	Brinco Ltd.	Asbestos	1990	15	10
Logtung	Amax	W, Mo	1990	15	13
Red Chris	Kidd Creek	Cu, Au	1990	20	7
Schaft Creek	Teck Corporation	Cu, Mo, Au, Ag	g 1992	25	80
Stikine Copper	Liard-Hudson Bay (?)	Cu, Ag, Au	2000	20	55
Sustut	Falconbridge	Cu	1996	8	<u> 10 </u>
				Sub-Total	<u>213</u>
Existing					
Cassiar	Brinco Ltd.	Asbestos	-	10	7
		TABLE 1.2	2		
	ELECTRICAI	L DEMAND FRO	DM MINING 1	rowns	
	Mine-Town	Workfor	ce	Dema	and (MW)
Adanac		350		1.	.6
	Cassiar	500		3	.0
	Eaglehead	250		1.	.1
	Kutcho Creek	150		0.	.7

350

250

250

750

450

300

1-3

1.6

i.1

1.1

3.4

2.0

1.4

133 MOUNT KLAPPAN OPERATIONS

The demand from the Mount Klappan operations arises from two main sources:

- (1) Power necessary for the mine operation to produce coal for sale. This would include the townsite.
- (2) Power necessary to produce sufficient coal to fuel the coalfired powerplant.

The power requirements for coal for sale have been obtained from a separate study for the mine operations. These values have been prorated to obtain the required power for providing coal for electrical generation. The coal tonnages required for the powerplants are presented in Part 3 of this study.

A summary of the Mount Klappan power requirements are provided in the following Table 1.3.

TABLE 1.3

MOUNT KLAPPAN OPERATIONS - POWER REQUIREMENTS

		Level A 1 000 000 tpy					Level B 5 000 000 tpy		
Size of Powerplant (MW)	<u>50</u>	<u>100</u>	<u>150</u>	<u>200</u>	<u>250</u>	100	<u>150</u>	<u>200</u>	<u>250</u>
(1) Power to Produce Coal for Sale (MW)	9	9	9	9	9	60	60	60	60
(2) Power to Produce Coal for Powerplant (MW)	3	5	7	_9	_11	5	7	_10	_12
Total Requirement (MW) =(1)+(2)	12	14	16	18	20	65	67	70	72

1-4

I.

1.4 LOAD GROWTH

141 INTRODUCTION

Expected start dates for development of major potential mines are very difficult to predict. To illustrate a possible maximum load growth pattern, an optimistic outlook on development of mines has been taken which assumes that all identified mines will have been started by the year 2000. The sequence of start dates has been established from information obtained informally and is believed to correspond to an optimistic development pattern adopted by the Provincial Government for its planning for the northwest of B.C.

142 LOAD GROWTH CURVES

Three curves have been prepared to illustrate load development using the information in Tables 1.1 and 1.2. These curves are for the following load areas, which are shown on Figure 1.1:

- Kutcho Creek Area
- Schaft Creek Area
- All areas.

The demands at Mount Klappan for the production of coal are listed on Figure 1.1, for both production levels and for each size of powerplant. These demands must be added to the curves to obtain total demand on a powerplant.

.1 Kutcho Creek Area

The following mines and towns have been included in this area:

- Red Chris
- Eaglehead
- Letain
- Kutcho Creek
- Dease Lake Town
- .2 Schaft Creek Area

The following mines and towns have been included in this area:

- Red Chris
- Schaft Creek
- Stikine Copper
- Eddontenajon and Telegraph Creek towns.

.3 All Areas

The following mines and towns have been included in these areas:

- Kutcho Creek area mines
- Schaft Creek area mines
- Dease Lake town
- Eddontenajon and Telegraph Creek towns
- Cassiar mine and town
- Adanac and Atlin town
- Logtung
- Sustut

1.5 YUKON AND ALASKA

Potential and existing mines in the Yukon and Alaska have not been included in this study. This decision was made following discussions with the B.C. and Yukon Chamber of Mines as to the potential for development in those areas compared with development in British Columbia.

In the Alaska panhandle, one prospect has been identified, but this is located on an island over 300 km from Mount Klappan. There are no large operating mines at present.

The exploration activity in the Yukon has involved expenditures of substantial amounts; nevertheless, development of substantial Yukon mines is not foreseen before the year 2000. The property of Amax/Logtung on the B.C./Yukon border has been included in the study.

Amax/Regional Resources have been actively exploring a number of lead-zinc deposits near Watson Lake (Midway deposits).

An area reported to have significant potential and exploration activity is around Macmillan Pass, however, this area is about 500 km north of Mount Klappan and straddles the Yukon/Northwest Territories border.

1.6 OTHER POTENTIAL MINES

The following areas or mines with identified potential have not been included in Table 1.1 but are mentioned here for reference.

1-6

ł

161 WINDY-CRAGGY

This property is owned by Falconbridge Ltd. and Geddes Resources Ltd. The property is located in the extreme north-west of B.C. near the Alsek River, about 200 km west of Atlin (Adanac mine area). Recent public announcements say that the "drill-indicated" tonnage (12 holes) is "100 million tons grading at 2.8 percent copper and 0.09 percent cobalt. Inferred tonnage for the overall deposit -- is more than 334 million tons with a grade of 1.52 percent copper and 0.08 percent cobalt." (A second report says that the latter figures were based on 3 drill holes).

162 STEWART/KITSAULT AREA

At present the Canada Wide Mine (Granduc) north of Stewart has power provided by diesel-fired gas turbine generators using bunker C oil. The mine has reported an expected life through to 1988, but is recognised as having to overcome difficult mining conditions. The diesel-electrical installation at Stewart run by BCHPA has a rated capacity of 7.25 MW. Peak demand in 1983 was around 3 MW, and the town depends heavily on activity at Granduc. There are other potential mines in the area, however, the difficulties of mining and access are considerable. If adequate access is provided, it may be practical to extend the transmission line from Kitsault. Similarly, in addition to Amax Kitsault, there are large potential properties in the Kitsault/Alice Arm area. Disposal of tailings could present significant environmental problems. Should potential properties in both the Stewart and Alice Arm areas be developed reasonably concurrently, a good possibility may exist for extension of BCHPA service from Kitsault.

163 MISCELLANEOUS PROPERTIES

The following properties in the area studied have been reported on over the last five to ten years.

- .1 New Jersey Zinc Exploration. Located near Cassiar, this property is reported as a potential producer. Reserves quoted in 1977 are 110 million tons with 0.115% Molybdenum. Expected life is approximately 15 years, based upon a 20,000 tpd production rate.
- .2 Gnat Pass

This Hudson Bay property is between Red Chris and Dease Lake, and was reported in 1974 as being a potential open-pit copper operation with an expected life of 10 years. It is relatively low grade ore, however, and has not been included in the study.

1.7 PRECIOUS METAL MINES

For this study the potential loads from the precious metal mines have not been included. A brief discussion follows.

Existing small precious metal mines are reported to be served by diesel-generator sets, often mobile units. Both existing and potential precious metal mines in the vicinity of major existing or proposed mineral developments may have some potential as future loads to be serviced from Mount Klappan powerplant. One such group is in the Cassiar area.

Economic connection of such loads to the transmission line from Mount Klappan would depend upon the evaluation of the following:

- size of load and expected life of mine;
- anticipated continuity of production;
- distance from the major substation and substation and transmission line costs to the mine;
- increased transformer and switching costs to existing mines if voltage at the mine is presently generated at the required voltage by the diesel-generator set without transformation;
- sharing of capital and operating costs of the major transmission system built into the rate proposed;
- effect on the small mine of system stability problems with the major mine loads on line;
- reliability of supply should load-shedding on the system be required to service a major mine load.

С



PART 2

 \bigcirc

 \bigcirc

COAL-FIRED ELECTRIC GENERATING STATIONS CAPITAL COST ESTIMATES

PART 2 - COAL-FIRED ELECTRIC GENERATING STATIONS CAPITAL COST ESTIMATES

î

CONTENTS

()

 \bigcirc

 \bigcirc

			Page
2.1	INTI	RODUCTION	2-1
2.2	SCO	OPE OF ESTIMATES	2-2
	221	Level A, 1 000 000 tonne per year	2-2
	222	Level B, 5 000 000 tonne per year	2-2
2.3	COA	AL QUALITY	2-3
	231	Specifications Used	2-3
		.l Level A, Proximate Analysis	2-3
		.2 Level B, Proximate Analysis	2-3
2.4	EQU	JIPMENT	2-3
	241	Boiler Plant Equipment	2-3
		.1 Steam Generator	2-3
		.2 Coal Handling System	2-3
		.3 Air Quality Control Systems	2-4
		.4 Ash and Sludge Disposal Systems	2-4
	242	Turbine-Generator and Related Systems	2-4
		.l Turbine-Generators	2-4
	243	Condenser Cooling Water System	2-4
	244	Switchyard Equipment	2-4
2.5	САР	PITAL COST ESTIMATES	
,>	251	Work Included	2_5
	252	Work and Costs Not Included	2-) 2 (
	474	HOLK GIG COSTS TAOL THOTAGED	2-0

			Page
	253	Basis of Estimates	2-6
		.l Labour	2-6
		.2 Material	2-6
		.3 Equipment	2-7
		.4 Construction Equipment	2-7
		.5 Taxes and Duty	2-7
		.6 Contractor's Overhead and Profit	2-7
		.7 Indirect Costs	2-7
	254	Capital Cost Curves	2-7
		.1 Station Size	2-7
		.2 Unit Costs	2-7
		.3 Varying Ash Content of Coal	2-8
2.6	СНС	DICE OF UNIT SIZE	2-8
2.7	COM	MENTS AND ALTERNATIVES	2-10
	271	Cooling Water Alternatives	2-10
		.1 Mechanical Draft Cooling Towers	2-10
		.2 Cooling Pond	2-10
		.3 Natural Draft Cooling Towers	2-11
		.4 Dry Type Radiator Systems	2-11
		.5 Wet/Dry Systems	2-11
	272	Sludge and Ash Disposal	2-11
	273	Fluidized Bed Combusion (FBC)	2-11
	274	Stikine Copper and Schaft Creek Mines	2-12
			· ·

Table

 \bigcirc

 \bigcirc

 \bigcirc

2.1	Twin Unit Installations, Capital Cost Estimate Summary - Level A
2.2	Twin Unit Installations, Capital Cost Estimate Summary - Level B

Figure

2.1 Unit Capital Costs - Twin Units Only

2.1 INTRODUCTION

This part covers order-of-magnitude capital cost estimates for nine sizes of coal-fired electric generating stations, ranging from 50 MW to 250 MW installed capacity. The station would be located at Mount Klappan in the general vicinity of the coal mine operations. Likely sites for the generating stations have not been identified.

The estimates have been based upon use of coal with ash contents and specifications given below. Comments are provided on the order-of-magnitude effects of varying ash contents on capital costs.

In Figure 2.1, unit cost curves are provided which show the relationship between the unit cost per kilowatt of installed capacity, for both levels of coal quality. The unit costs shown are based upon twin-unit installations.

Factors have been provided in Section 2.6 which may be used with the cost curves to obtain cost estimates of stations employing various numbers and sizes of units. These combinations can be chosen to allow for variations in load buildup schedules and for variations in assumptions regarding standby units required for operation during periods of maintenance and down-time of other units.

Comments are included on alternatives for cooling of condensate water, and relative cost figures are provided.

The matters of availability of water, means of ash disposal, and local environmental considerations can have a significant influence on the design of the associated facilities at the plant and their corresponding capital and operating costs. Investigation of these matters is beyond the scope of this study.



2.2 SCOPE OF ESTIMATES

The estimates have been prepared for two levels of coal production, designated in this study as Level A and Level B, with two different ash contents, namely, 33% ash for Level A and 20% ash for Level B.

221

Level A: 1 000 000 tonnes per year of run-of-mine coal shipped as product. Powerplants would use run-of-mine coal which would be screened for oversize only prior to delivery to the plant. The sizes of the stations are:

.1 50 MW net station output
.2 100 MW net station output
.3 150 MW net station output
.4 200 MW net station output
.5 250 MW net station output

222

Level B: 5 000 000 tonnes per year of washed product coal. Power plants would use washed coal from the coal preparation plant. The sizes of the stations are:

- .1 100 MW net station output
- .2 150 MW net station output
- .3 200 MW net station output
- .4 250 MW net station output

2.3 COAL QUALITY

231 SPECIFICATIONS USED

Following are the coal specifications adopted for this study:

.l Level A – Proximate Analysis

Residual Moistur Ash Volatile Matter Fixed Carbon Total Sulphur	e - - - -	1.59 32.29 6.69 59.79 0.79	6 6 6 6
нν	-	21.36	6 MJ/kg
	- 5	5116	kcal/kg

.2 Level B - Proximate Analysis

Residual Moisture -		1.6%			
Ash	19.0%				
Volatile Matter	-	8.3	ж		
Fixed Carbon	-	71.19	ж		
Total Sulphur	-	0.6	ж		
HV	-	27.2	MJ/kg		
	-	6506	kcal/kg		

2.4 EQUIPMENT

The following provides a brief description of the major components of the powerplant equipment.

241 BOILER PLANT EQUIPMENT

.1 Steam Generator

Type - indoor, pulverized coal, balanced draft. Pressure/Temperature - to match turbine generator requirement.

.2 Coal Handling System

The coal is to be transported from the coal mine to the powerplant site. The transportation system is not included. Coal delivery to the plant site is assumed to be daily except Saturday and Sunday. The coal distribution to stockpile is to be accomplished in one shift - 6 hours operation.

The conveying system will be complete with dry dust collection, wet dust suppression, sampling, crushing house, 2-3

transfer tower, tramp iron magnets, belt scale, stationary trigger conveyor, feeders, hoppers, and boiler feed bins.

.3 Air Quality Control Systems

The boilers will be equipped with electrostatic precipitator for fly ash emission control and wet-type limestone scrubber for control of sulphur dioxide emissions.

.4 Ash and Sludge Disposal Systems

A pneumatic conveying system will remove ash from the boilers. Vacuum will be created by a mechanical exhauster. Ash will be stored in a silo and conditioned for dust control and unloaded at regular intervals. Sludge will be conditioned with fly-ash for ease of handling and removed at the regular intervals to the sludge disposal area.

242 TURBINE - GENERATOR AND RELATED SYSTEMS

.1 Turbine-Generators

The steam cycles assumed for the units are as follows:

- . Twin Unit-27 MW (50 MW, net) 1250 psig/950°F, four feedwater heaters 430°F feed temperature
- . Twin Unit-54 MW (100 MW, net) 1250 psig/950°F, four feedwater heaters, 430°F feed temperature
- . Twin Unit-80 MW (150 MW, net) 1800 psig/1000°F, five feedwater heaters, 450°F feed temperature
- . Twin Unit-107 MW (200 MW, net) 1800 psig/1000°F, six feedwater heaters, 450°F feed temperature
- . Twin Unit-134 MW (250 MW, net) 1800 psig/1000°F, six feedwater heaters, 450°F feed temperature
- <u>NOTE</u>: A comment on the use of other size units or installing standby units for reliability of supply and the effect on the initial capital investment is presented in Section 2.5.

243 CONDENSER COOLING WATER SYSTEM

The system will employ evaporative type, mechanical draft cooling tower of wooden construction. A comment on the use of a cooling pond and other systems as alternatives to the cooling tower scheme is presented in Section 271. The source of raw water has not been identified for this study.

244 SWITCHYARD EQUIPMENT

The Switchyard will include all necessary equipment ready for connection of high voltage transmission lines.

2.5 CAPITAL COST ESTIMATES

A summary of the order-of-magnitude cost estimates for the various sizes of twin-unit powerplant for Level A and Level B are presented in Table 2.1. The range of costs have been presented in graphical form in Figure 2.1. As explained in Section 2.6, the graph and factors presented can be used to estimate the cost of a spare unit or units to provide a given firm station capacity for the load characteristics assumed.

251 WORK INCLUDED

Following are the major items of work included in the estimates:

- .1 Boiler Plant Equipment and Related Systems: As outlined in Section 241.
- .2 Turbine Generators and Related Systems: As outlined in Section 242.
- .3 Electrical Plant Systems and Equipment.
- .4 Miscellaneous Powerplant Equipment, including spares, startup oil and materials.
- .5 Switchyard equipment complete ready for connection of high-voltage transmission lines.
- .6 Buildings and structures, including main power building, administration building, warehouse and machine-shop, gate-house; stack.
- .7 Raw Water Supply. An allowance has been included for this system. No work has been done, however, to identify the potential location, type and pumping systems necessary to supply raw water to the powerplants.
- .8 On-Site Ground Improvements, including site for coal pile, clearing, excavation and fill, roads, fencing, drainage, sewage and fire and domestic water systems, yard lighting, all within the battery limits only. A reasonably level site has been assumed with no unusual foundation conditions.
- .9 Sludge Disposal. The location and type of sludge disposal area depend upon a detailed study of site conditions and environmental requirements. An allowance only has been included for the disposal area, and it is emphasized that actual costs can vary widely.
- .10 Camp Costs. An allowance has been made for camp costs during construction.

.11 Coal in Storage. An initial supply of coal in storage sufficient for 6 months operation has been included in the estimates.

252 WORK AND COSTS NOT INCLUDED

The following items have not been included in the cost estimates.

- .1 Access. Road and rail access to the powerplant site have been assumed to be available.
- .2 Land Costs.
- .3 Offsite Improvements.
- .4 Interest During Construction.
- .5 Owner's Costs, both during construction and for housing of operating staff.
- .6 Construction Model.
- .7 Mobile Equipment for Coal Supply and Ash Disposal.
- .8 District Heating for local town. Note: An opportunity exists at Mount Klappan for district heating and should be studied further.
- .9 Escalation.

253 BASIS OF ESTIMATES

The cost estimates are based on 1982 costs in Canadian dollars. No drawings were prepared and no specific site location studies were conducted.

Costs contained in Table 2.1 are based upon the following elements.

.l Labour

Labour costs include labour, payroll burden, travel costs, consumables and small tools, and contractor's overhead and profit. A 45 hour work week (37% hours straight time and 7% hours double time) has been assumed as an incentive to attract a skilled work force.

.2 Material

Material pricing was based on in-house information and experience on projects in British Columbia. All material costs are assumed to be F.O.B. jobsite.

.3 Equipment

For equipment pricing, budget telephone quotes were obtained for the steam generators, electrostatic precipitators, scrubbers, major pumps, turbine generators and condensers. Other equipment was priced from in-house data.

.4 Construction Equipment

Equipment usage is included in the unit cost and is the cost of construction equipment excluding operator and oiler.

.5 Taxes and Duty

An allowance of 5% has been included for these items.

.6 Contractor's Overhead and Profit

Contractor's overhead and profit is included in the unit prices.

.7 Indirect Costs

Other indirect costs such as professional services and construction management are calculated at 15% of construction cost plus spare parts and lube oil/start materials, etc.

254 CAPITAL COST CURVES

The Figure 2.1 curves show the order-of-magnitude relationship between station size for the unit combinations shown and the unit cost in dollars per kilowatt installed. Some particular comments follow.

.1 Station Size

The unit costs have been based upon installation of twin units of output shown. The installed capacities are shown in parentheses. The difference between the two corresponds to an assumed station service power requirement of 7%.

The curves are intended as an aid in determining the relative costs of twin coal-fired units when used in combination with other coal-fired units, or diesel electric units, for standby (necessary to provide the firm capacity of the station). The costs of the standby units can be added to the twin unit costs as explained in Section 2.6.

.2 Unit Costs

The unit costs of stations below the 100 MW capacity rise very rapidly, and the values can be quite irregular depending 2-7

upon size combinations and specifications chosen. For this reason, the curves have not been drawn between values shown for stations sized below 100 MW. A rough unit cost for a 30 MW station would be \$2,900 per kilowatt installed for Level A coal, assuming twin 16 MW units.

.3 Varying Ash Content of Coal

The two curves shown for Level A and Level B coals indicate the difference in unit costs for the two ash levels. Boiler manufacturers quoted a range of 10% to 15% increase in the boiler plant cost for a corresponding increase in coal ash content from 20% to 33%. The cost of the balance of the plant remains the same. Interpolation between these curves for ash contents between 20% and 33% will provide a reasonable unit cost estimate.

2.6 CHOICE OF UNIT SIZE

The cost estimates contained in Table 2.1 have been based upon the installation of twin units to provide the installed capacities shown.

These cost estimates of twin units have been prepared to form a basis or framework from which to estimate the costs of a powerplant in which various combinations of unit sizes are chosen to suit load assumptions, demand variations, and standby units required for operation during periods of overhaul and downtime on other units.

Example 1: For a firm capacity of 100 MW the powerplant could consist of 3-50 MW units installed capacity, with one unit as standby. Another choice may be 2-50 MW units and 2-25 MW units installed, which would provide more flexibility in operation to accommodate varying demands and downtime, and can afford more efficient operating ranges in the larger units by use of one of the smaller units at full load in place of a second larger unit at half load.

Order-of-magnitude comparisons can be made in the same way for powerplant sizes chosen to suit a given combination of mine loads, and adjustments can be made to assess the effects of differing demand levels from mines and various assumptions as to sequence of development and anticipated life of mines.

Similarly, order-of-magnitude comparisons can be made between smaller, coal-fired standby units and comparable diesel generator units, the costs of which are set out in Part 5.

For convenience, as explained in Section 254, the curves given in Figure 2.1 can be used to make these comparisons together with the application of the following factors which are based upon experience. This may be best illustrated by further examples, as follows:

Example 2: <u>Three-Unit Installation, with one unit used as a standby.</u>

To estimate the cost per kilowatt of the standby unit, obtain the cost per kilowatt installed (for twin units) from Figure 2.1, then multiply the result by 85%.

50 MW net firm station output. Choose 3 x 27 MW units = 81 MW installed capacity, using one unit as a standby unit.

	•
Twin Units - unit cost Spare Unit - unit cost	= \$2,790/kW installed = 0.85 x \$2,790 = \$2,370/kW installed
Total installed cost	= \$2,790/kW x 54 x 10 ³ kW + \$2,370/kW x 27 x 10 ³ kW
	= \$215 Million
* 81 MW	= \$2,655/kW installed

Example 3:

Four-Unit Installation, with a diesel unit used as standby

To estimate the cost per kilowatt of four similar coal-fired units, obtain the cost per kilowatt installed (for two units) from Figure 2.1, then multiply the result by 95%.

100 MW net firm station Choose 4 x 27 MW coal-fired units	n output.
plus 25 MW diesel	
standby capacity	= 133 MW installed capacity.
Four Units – unit cost	= 0.95 x \$2,790/kW = \$2.650/kW installed
Diesel Units – unit cost (Table 5.1)	= \$775/kW installed
Total Installed Cost	= \$2,650 x 108 x 10 ³ + \$775 x 25 x 10 ³ = \$305 Million
+133 MW	= \$2,295/kW installed

Use of unit costs of diesel units for standby capacity as given in Part 5 will result in a conservative estimate of overall cost because the coalfired station costs include an allowance for black-start diesel capability. 2-9 The effect on the unit costs is not large and is within the order-ofmagnitude of the estimates.

2.7 COMMENTS AND ALTERNATIVES

271 COOLING WATER ALTERNATIVES

No investigation was made in this study of potential sites and sources for obtaining raw water for the powerplants, and no investigation was made to identify potential sites for cooling ponds. Similarly, no attempt was made to evaluate possible environmental effects of various kinds of cooling towers or ponds.

.1 Mechanical Draft Cooling Towers

The cost estimates are based upon the use of this type of cooling tower. The approximate water requirement for domestic use, cooling tower makeup, boiler water makeup, and other miscellaneous services for both Level A and Level B are as follows:

Plant Size, MW	50	100	150	200	250

Water Requirement, US g.p.m. 2100 4100 5600 6700 8100

<u>NOTE</u>: Approximately 96% of the above water requirement is for cooling tower makeup.

The effects of possible formation of fog from the cooling tower must be evaluated for any site location and climatic conditions as to effects on the actual operations and on the environment in general.

.2 Cooling Pond

In general, the initial investment cost of a cooling pond could be expected to be about 150% higher than the cost of mechanical draft cooling towers. However, the operating costs associated with the cooling pond could be much lower and the pond aesthetically preferred over the cooling tower scheme. The difference in operating costs would depend upon the location of the pond. The environmental impact or effect of the cooling pond would also be less and contained in a smaller area than the cooling tower scheme. In addition, possibly the cooling pond could also be utilized for recreational purposes.
Based on climatological conditions at the site, the cooling pond area requirements for the units would be approximately 0.70 acre to 1.0 acre for each installed megawatt plant capacity or about 14 acre feet per MW.

.3 Natural Draft Cooling Towers

These are about twice as expensive to build as mechanical draft towers.

.4 Dry Type Radiator Systems

These air cooled systems (direct and indirect) cost about three and a half to four times as much to build as a mechanical cooling tower system and require large volumes of air for cooling. Potential acoustical problems must be considered with these systems.

.5 Wet/Dry Systems

The systems operate wet in summer (rely on water evaporation for cooling) and dry in winter. They cost about twice as much as the mechanical draft systems.

272 SLUDGE AND ASH DISPOSAL

No study has been made as to potential sludge and ash disposal sites. The environmental climatic and geological considerations can have a marked influence on the choice of a site and the type of disposal area and relationship to the mining operation, and must be studied in detail for each specific site. Disposal of sludge in winter presents substantial storage, transporting and disposal problems and, similarly, must be the subject of detailed study.

273 FLUIDIZED BED COMBUSTION (FBC)

Recent budget pricing information of an FBC boiler of the atmospheric type shows competitiveness with conventional boilers.

Some of the major advantages are as follows:

- Expected lower operating and maintenance costs.
- Elimination of energy requirements to operate flue gas desulphurization system (FGD).
- Sulphur dioxide (SO₂) removal efficiency is simply adjusted for different coals by changing the coal to limestone ratio.
- Lower production of oxides of nitrogen (NO_X) because of reduced combustion temperatures.

2-11

- Spent limestone is removed with the ash in a dry, easily stored form (unlike the large volumes of wet FGD sludge which are difficult to transport and store).
- Unburned carbon is passed through a carbon burn-up cell (CBC) for complete combustion to remove monoxide, unburned hydrocarbons and soot.

The potential advantages of the fluidized bed combustion system are so significant that it is strongly recommended a study be made of comparative capital and operating costs. Evaluation of the environmental advantages of the system should also be included in the study.

274 STIKINE COPPER AND SCHAFT CREEK MINES

The demands from both of these mines would be very high, and that from Stikine Copper would be higher than any existing open-pit mine in British Columbia. The influence of either one or both of these mines, and the expected start dates, on the sizes of units chosen in a powerplant would be most significant.

2-12

TABLE 2.1

CAPITAL COST ESTIMATE SUMMARY

COAL-FIRED ELECTRIC GENERATING STATIONS

(\$000 Cdn - 1982)

		LEVEL A -	I Million Tonne P	er Year					
DESCRIPTION	50 M₩ (2 x 27 M₩)	100 MW (2 x 54 MW)	150 MW (2 x 80 MW)	200 MW (2 x 107 MW)	250 MW (2 x 134 MW)				
ON-SITE FACILITIES									
On-site Ground Improvements Buildings and Structures Boiler Plant Equipment	3,250 14,700 47,400	3,700 15,350 87,100	4,100 17,700 131,900	4,500 20,800 159,250	5,150 24,200 184,250				
Electrical Plant Systems & Equipment Miscellaneous Power Plant Equipment	6,800 2,600 2,200	11,000 4,450 3,350	16,100 6,050 4,350	20,650 7,800 5,700	63,200 23,750 9,050 6,900				
TOTAL DIRECT CONSTRUCTION COST	\$ 94,650	152,150	217,200	276,500	318,500				
Taxes and Duty Spare Parts Allowance Lube Oil/Start Materials etc.	6,450 960 40	10,350 1,500 50	14,780 2,150 70	18,800 2,710 90	21,650 3,140 110				
SUB-TOTAL	\$102,100	164,050	234,200	298,100	343,400				
Professional Services Engineering Services and Procurement Construction Management	15,400	24,600	35,100	44,700	51,500				
SUB-TOTAL	\$117,500	188,650	269,300	342,800	394,900				
Contingency	23,500	37,750	53,900	68,400	78,900				
TOTAL ON-SITE COST	\$141,000	226,400	323,200	411,200	473,800				
OFF-SITE FACILITIES									
Camp Cost Raw Water Supply Ash/Sludge Disposal Area	6,600 800 500	10,650 1,400 800	15,200 1,900 1,000	19,350 2,300 1,200	22,300 2,600 1,400				
SUB-TOTAL	\$ 7,900	12,850	18,100	22,850	26,300				
Contingency	1,600	2,550	3,600	4,550	5,300				
TOTAL OFF-SITE COST	<u>\$ 9,500</u>	15,400	21,700	27,400	31,600				
TOTAL CAPITAL COST	150,500	241,800	344,900	438,600	505,400				
Initial Coal Supply - 6 months	6,600	13,200	19,200	25,900	32,000				

(Level A, \$50.00/t. Level B, \$75.00/t)

Caution: The above capital costs must be adjusted for costs of standby capacity. See Section 2.6.

TABLE 2.2

CAPITAL COST ESTIMATE SUMMARY

COAL-FIRED ELECTRIC GENERATING STATIONS

(\$000 Cdn - 1982)

LEVEL B - 5 Million Tonne Per Year

DESCRIPTION	100 M₩ (2 x 54 M₩)	150 MW (2 x 80 MW)	200 M₩ (2 x 107 M₩)	250 M₩ (2 x 134 M₩)
ON-SITE FACILITIES				
On-site Ground Improvements Buildings and Structures Boiler Plant Equipment Turbine Generator & Related Systems Electrical Plant Systems & Equipment Miscellaneous Power Plant Equipment Switchyard	3,500 13,600 78,500 27,200 11,000 4,100 3,350	3,650 15,800 118,050 37,000 17,000 6,050 4,350	3,900 18,600 142,650 57,800 20,650 7,800 5,700	4,500 21,650 164,950 65,200 23,750 9,050 6,900
TOTAL DIRECT CONSTRUCTION COST	141,250	201,900	257,100	296,000
Taxes and Duty Spare Parts Allowance Lube Oil/Start Materials etc.	9,600 1,410 40	13,650 1,990 60	17,500 2,530 70	20,100 2,910 90
SUB-TOTAL	152,300	217,600	277,200	319,100
Professional Services Engineering Services and Procurement Construction Management	22,900	32,500	41,600	47,800
SUB-TOTAL	175,200	249,100	318,800	366,900
Contingency	35,000	49,800	63,800	73,400
TOTAL ON-SITE COST	210,200	298,900	382,600	440,300
OFF-SITE FACILITIES				
Camp Cost Raw Water Supply Ash/Sludge Disposal Area	9,900 1,400 600	14,050 1,900 800	18,000 2,300 1,000	20,700 2,600 1,100
SUB-TOTAL	11,900	16,750	21,300	24,400
Contingency	2,400	3,250	4,200	4,900
TOTAL OFF-SITE COST	14,300	20,000	25,500	29,300
TOTAL CAPITAL COST	224,500	318,900	418,100	469,600
Initial Coal Supply - 6 months	16,000	22,400	31,000	38,000

(Level A, \$50.00/t. Level B, \$75.00/t)

Caution: The above capital costs must be adjusted for costs of standby capacity. See Section 2.6.

PART 3

6

 \bigcirc

 \bigcirc

COAL-FIRED ELECTRIC GENERATING STATIONS

OPERATING COST ESTIMATES

PART 3 - COAL-FIRED ELECTRIC GENERATING STATIONS OPERATING COST ESTIMATES

Page

CONTENTS

3.1	INTI	RODUCTION	3-1
3.2	BAS	IS OF ESTIMATES	3-1
	321	Cost of Coal	3-1
	322	Sludge/Ash Removal	3-1
	323	Operating and Maintenance Materials and Chemicals	3-2
	324	Operating and Maintenance Labour	3-2
	3 25	Management and Supervision	3-2
	326	Incremental Cost of Fuel Oil	3-2
	327	Station Capacity Factor	3-2

Table

3.1	Annual Operating Cost Estimates
	Coal-Fired Electric Generating Stations
	Coal Cost: Level A - \$50.00/tonne; Level B - \$75.00/tonne

- 3.2 Annual Operating Cost Estimates Coal-Fired Electric Generating Stations Coal Cost: \$25.00/tonne
- 3.3 Annual Operating Cost Estimates Coal-Fired Electric Generating Stations Coal Cost: \$75.00/tonne

3.1 INTRODUCTION

The annual operating cost estimates for the nine coal-fired generating stations from 50 MW to 250 MW capacity are presented in Tables 3.1, 3.2 and 3.3. Each table is based upon different assumptions for cost of coal.

The costs shown include the cost of fuel, labour, plant management, operating and maintenance materials and services, and sludge/ash disposal.

For Level A, 1 000 000 tonne per year production of product coal, the operating costs have been based upon use of coal with a 33% ash content as discussed in Part 2.

The coal costs are based upon the tonnage required to produce the net output capacity assuming that 100% of the output would be required 100% of the time (8760 hours per year) using coal only as fuel.

The actual installed capacity necessary to provide this output would be larger and the spare or standby capacity necessary could be provided by coal-fired units or, more probably, by diesel generator sets or oil-fueled gas turbine-generator sets.

3.2 BASIS OF ESTIMATES

321 COST OF COAL

The required annual tonnage of coal is based upon coal providing 100% of the output. The unit costs per tonne have been assumed to be:

	Level A, 1 000 000 tpy	Level B, 5 000 000 tpy
Table 3.1	\$50.00	\$75.00
Table 3.2	\$25.00	\$25.00
Table 3.3	\$50.00	\$50.00

As can be seen from the tables, the operating costs per kWh are most sensitive to the price of coal for all assumed unit costs of coal.

322 SLUDGE/ASH REMOVAL

A unit cost of \$2.00 per tonne has been assumed. As discussed in Part 2, the unit cost will depend upon actual site conditions and the effects of winter weather on storage, handling and transportation facilities.

3-1

323 OPERATING AND MAINTENANCE MATERIALS AND CHEMICALS

Estimates for these items have been based upon data available from similar installations.

324 OPERATING AND MAINTENANCE LABOUR

The numbers of operators and maintenance personnel required have been based upon data available from similar sized plants. The average cost per year has been taken as \$47,000 per man, including fringe benefits.

325 MANAGEMENT AND SUPERVISION

The cost per year for these personnel for the numbers estimated has been based upon an annual salary cost of \$65,000 per person, including fringe benefits.

326 INCREMENTAL COST OF FUEL OIL

Should oil-fueled diesel-generator or gas turbine-generator sets be used for standby service rather than a similar size coal-fired unit, it would be necessary to add to the operating costs shown in Table 3.1, 3.2 and 3.3, the incremental cost of fuel oil used.

The amount of fuel-oil used would depend upon the combination of unit sizes chosen for a given station capacity and the assumed station capacity factors for given load combinations. This amount of fuel oil would be in addition to that required for boiler start-up and for combustion stabilization when a unit may be operating at low load conditions.

An order-of-magnitude comparison of the fuel cost in cents per kWh is:

	Table 3.1	<u>Table 3.2</u>	Table 3.3
- Raw Coal:	3.0	1.5	3.0
- Washed Coal:	3.5	1.2	2.4
- Diesel Oil:	10.0	10.0	10.0

327 STATION CAPACITY FACTOR

In order to provide the firm power output on which the tables have been based, it probably would be more practical and economical to install oil-fueled gas turbine-generator or diesel generator sets rather than a similarly sized coal-fired unit for backup or standby service. The amount of fuel oil used would depend upon the assumed capacity factor of the station for servicing mine loads and the combination of sizes of units chosen to economically serve the system. For comparison, the typical lifetime capacity factor of 3-2 utility type coal-fired powerplants of the unit sizes and technical features assumed in the study is approximately 70%. (This figure is based upon the Edison Electric Institute (EEI) annual survey of operating powerplants in the U.S.A.).

The expected overall economy of utilizing oil-fired standby units would result from lower capital costs of the sets compared with coal-fired units. The operating costs could be higher, however, this would depend upon the load levels at which the coal-fired units would be operating.



 \bigcirc

 \bigcirc

 \bigcirc

Coal Cost Level A - \$50.00/tonne Level B - \$75.00/tonne

.

0.

Level B - \$75.00/tonne			LEVEL A - 1 M	tpy		LEVEL B – 5 Mtpy			
DESCRIPTION	50 MW	100 MW	150 MW	200 MW	250 MW	100 MW	150 MW	200 MW	250 MW
Materials and Miscellaneous									·
Coal Requirement-t/yr Ash/Sludge Flow-t/yr	(264,000) (130,000)	(526,000) (260,000)	(754,000) (375,000)	(1,034,000) (510,000)	(1,280,000) (625,000)	(422,000) (125,000)	(596,000) (175,000)	(824,000) (240,000)	(1,018,000) (295,000)
Coal Cost Ash/Sludge Disposal	13,200 260	26,300 520	37,700 750	51,700 1,020	64,000 1,250	31,600 250	44,700 350	61,800 480	76,300 590
Operating Materials, Chemicals	190	260	320	380	450	250	300	350	400
Maintenance Materials, Services	550	1,000	1,500	2,000	2,500	900	1,350	1,870	2,300
Sub-Total	14,200	28,100	40,300	55,100	68,200	33,000	46,700	64,500	79,600
Labour									
Operating, Maintenance, Misc. Cost @ \$47,000/yr	(57) 2,680	(75) 3,520	(95) 4,470	(105) 4,930	(115) 5,400	(75) 3,520	(95) 4,470	(105) 4,930	(115) 5,400
Manager, Supervisory Cost @ \$65,000/yr	(6) 390	(7) 450	(8) 520	(9) 590	(9) 590	(7) 450	(8) 520	(9) 590	(9) 590
Sub-Total	3,100	4,000	5,000	5,500	6,000	4,000	5,000	5,500	6,000
TOTAL ANNUAL COST	17,300	32,100	45,300	60,600	74,200	37,000	51,700	69,000	85,600
Annual Energy Output-kWh x 106	438	876	1,314	1,752	2,190	876	1,314	1,752	2,190
Operating Cost-¢/kWh	3.95	3.66	3.45	3.46	3.39	4.22	3.93	3.94	3.91

Note: The annual operating cost estimates are based upon 100% load operation, 24 hours per day, 365 days per year = 8,760 kWh/year produced for each kW net output. The unit cost per kWh energy output is very sensitive to assumed cost per tonne of coal. Totals have been rounded.

TABLE 3.1

ANNUAL OPERATING COST ESTIMATES

COAL-FIRED ELECTRIC GENERATING STATIONS

(\$ 000 Cdn - 1982)

TEVEL B S Menu



 \bigcirc

()

 \bigcirc

Coal Cost

- \$25.00/tonne

DESCRIPTION 50 MW 100 MW 150 MW 200 MW 250 MW 100 MW 15 Materials and Miscellaneous Coal Requirement-t/yr (264,000)(526,000)(754,000)(1,034,000)(1,280,000)(422,000)(59 Ash/Sludge Flow-t/yr (130,000)(260,000)(375,000)(510,000)(125,000)(625,000)(17 Coal Cost @ \$25.00/tonne 6,600 13,100 18,800 25,800 32,000 10,500 Ash/Sludge Disposal 260 520 750 1,020 1,250 250 Operating Materials, Chemicals 190 260 320 380 450 250 Maintenance Materials, Services 550 1,000 1,500 2,000 2,500 900 Sub-Total 7,600 15,000 21,500 29,300 36,200 11,900 Labour Operating, Maintenance, Misc. Cost @ \$47,000/yr (57) (75) (95) (105) (115) (75) 2,680 3,520 4,470 4,930 5,400 3,520 Manager, Supervisory (6) (7) (8) (9) (9) (7) Cost @ \$65,000/yr 390 450 520 590 590 450 Sub-Total 3,100 4,000 5,000 5,500 6,000 4,000 TOTAL ANNUAL COST 10,700 19,000 26,500 34,800 42,200 15,900 2 Annual Energy Output-kWh x 106 438 876 1,314 1,752 2,190 876 Operating Cost-¢/kWh 2.44 2.17 2.02 1.99 1.93 1.82

LEVEL A - 1 Mtpy

Note: The annual operating cost estimates are based upon 100% load operation, 24 hours per day, 365 days per year = 8,760 kWh/year produced for each kW net output. The unit cost per kWh energy output is very sensitive to assumed cost per tonne of coal. Totals have been rounded.

TABLE 3.2

ANNUAL OPERATING COST ESTIMATES

COAL-FIRED ELECTRIC GENERATING STATIONS

(\$ 000 Cdn - 1982)

LEVEL B - 5 Mtpy

0 MW	200 MW	250 MW
96,000) 75,000)	(824,000) (240,000)	(1,018,000) (295,000)
14,900 350	20,600 480	25,500 590
300	350	400
1,350	1,870	2,300
16,900	23,300	28,800
(95) 4,470	(105) 4,930	(115) 5,400
(8) <u>520</u>	(9) 590	(9) 590
5,000	5,500	6,000
21,900	28,800	34,800
1,314	1,752	2,190
1.67	1.64	1.59



 \bigcirc

 \bigcirc

 \bigcirc

Coal Cost - \$50.00/tonne

C

DESCRIPTION	50 MW	100 MW	150 MW	200 MW	250 MW	100 MW	150 MW	200 MW	250 MW
Materials and Miscellaneous				<u> </u>	·····		Î		
Coal Requirement-t/yr Ash/Sludge Flow-t/yr	(264,000) (130,000)	(526,000) (260,000)	(754,000) (375,000)	(1,034,000) (510,000)	(1,280,000) (625,000)	(422,000) (125,000)	(596,000) (175,000)	(824,000) (240,000)	(1,018,000) (295,000)
Coal Cost Ash/Sludge Disposal	13,200 260	26,300 520	37,700 750	51,700 1,020	64,000 1,250	21,100 250	29, <mark> </mark> 800 350	41,200 480	50,900 590
Operating Materials, Chemicals	190	260	320	380	450	250	300	350	400
Maintenance Materials, Services	550	1,000	1,500	2,000	2,500	900	1,350	1,870	2,300
Sub-Total	14,200	28,100	40,300	55,100	68,200	22,500	31,800	43,900	54,200
Labour									
Operating, Maintenance, Misc. Cost @ \$47,000/yr	(57) 2,680	(75) 3,520	(95) 4,470	(105) 4,930	(115) 5,400	(75) 3,520	(95) 4,470	(105) 4,930	(115) 5,400
Manager, Supervisory Cost @ \$65,000/yr	(6) <u>390</u>	(7) 450	(8)	(9) 590	(9) 590	(7) 450	(8) 520	(9) 590	(9) 590
Sub-Total	3,100	4,000	5,000	5,500	6,000	4,000	5,000	5,500	6,000
TOTAL ANNUAL COST	17,300	32,100	45,300	60,600	74,200	26,500	36,800	49,400	60,200
Annual Energy Output-kWh x 106	438	876	1,314	1,752	2,190	876	1,314	1,752	2,190
Operating Cost-¢/kWh	3.95	3.66	3.45	3.46	3.39	3.03	2.80	2.82	2.75

LEVEL A - 1 Mtpy

Note: The annual operating cost estimates are based upon 100% load operation, 24 hours per day, 365 days per year = 8,760 kWh/year produced for each kW net output. The unit cost per kWh energy output is very sensitive to assumed cost per tonne of coal. Totals have been rounded.

TABLE 3.3

ANNUAL OPERATING COST ESTIMATES COAL-FIRED ELECTRIC GENERATING STATIONS (\$ 000 Cdn - 1982)

LEVEL B - 5 Mtpy

PART 4 ŀ

)

Table

- 4.1 Capital Cost Estimate, Transmission System 1
 4.2 Capital Cost Estimate, Transmission System 2
 4.3 Capital Cost Estimate, Transmission System 3
 4.4 Capital Cost Estimate, Transmission System 4
 4.5 Capital Cost Estimate, Transmission System 5
 4.6 Capital Cost Estimate, Transmission System 6
 4.7 Capital Cost Estimate, Transmission System 7
 4.8 Capital Cost Estimate, Transmission System 8
- 4.9 Transmission Systems, Operating Cost Estimates

Plate

- 4.1 Transmission System 1
- 4.2 Transmission System 2
- 4.3 Transmission System 3
- 4.4 Transmission System 4
- 4.5 Transmission System 5
- 4.6 Transmission System 6
- 4.7 Transmission System 7
- 4.8 Transmission System 8

4.1 INTRODUCTION

Estimates of capital costs of transmission line systems from Mount Klappan of 138 kV and 230 kV capacity, and of switching and distribution substations, are presented in Table 4.1 through Table 4.8. These systems would serve major potential loads from large mines and townships which have been identified in Part 1 and shown on Plate 1.1.

The operating cost estimates are presented in Table 4.9.

All systems have been based on the provision of single circuit transmission lines, and no grid has been assumed for any system.

Various alternative systems have been identified for preparation of cost estimates. These systems have been chosen to serve various groups of loads from major potential mines. The systems chosen are shown on Plates 4.1 through 4.8. The choice of voltage for the systems has been based upon empirical rules-of-thumb and discussions with BCHPA.

4.2 SYSTEM ALTERNATIVES

In order to provide comparisons between costs of various transmission and associated substation systems which could serve the major load areas identified, a series of alternative systems have been chosen to reflect the possible development of the two major load areas as occurring at significantly different times, namely, Kutcho Creek area and Schaft Creek area. Sufficient combinations are presented in the alternative systems studied to allow order-of-magnitude comparisons within parts or sections of various systems.

For convenience of reference, for each system a map is provided which shows the mines served and transmission route (Plates 4.1 through 4.8). A table which shows the components of the system and the capital cost estimate for same follows each map.

The systems chosen have been divided broadly into 138 kV and 230 kV systems. Following are the systems for which estimates have been made:

138 kV Systems

System 1 (Plate 4.1) Mount Klappan to Red Chris Mine and Eddontenajon town, thence northeast to Eaglehead/Letain/Kutcho Creek mines.

System 2 (Plate 4.2)	Mount Klappan to Red Chris Mine, thence north to Dease Lake, and thence east to Eaglehead/Letain/Kutcho Creek mines.					
System 3 (Plate 4.3)	System 2 plus Dease Lake to Cassiar and thence west to Atlin (Adanac Mine).					
System 4 (Plate 4.4)	Mount Klappan to Sustut Mine to the southeast.					
230 kV Systems						
System 5 (Plate 4.5)	Mount Klappan to Red Chris Mine, Telegraph Creek and thence south to Stikine Copper and Schaft Creek Mines.					
<u>System 6</u> (Plate 4.6)	System 5 plus Dease Lake, Eaglehead/Letain/Kutcho Creek mines.					
<u>System 7</u> (Plate 4.7)	All loads.					

System 8Alternative route west to Stikine Copper and Schaft(Plate 4.8)Creek Mines via Bob Quinn Lake.

4.3 CAPITAL COST ESTIMATES

431 BASIS OF ESTIMATES

The capital cost estimates have been based generally upon discussions with BCHPA who have made studies over the years on potential transmission corridors and have assessed the general physical conditions of the potential areas to be served.

432 ROUTE SELECTION

The routes for each of the system alternatives selected have been based upon information provided by BCHPA from their studies assuming Mount Klappan as the power source.

433 LINE LENGTH

Lengths of transmission lines have been estimated from small scale maps, and are of order-of-magnitude accuracy. Each system is based upon terminating the line at the mine named at the end of the system.

4-2

Ī

434 LINE COSTS

Unit costs for 128 kV and 230 kV line construction were based upon discussions with BCHPA who have performed route studies and cost studies for transmission systems in the area from Mount Klappan to potential mines. These prices were subsequently reviewed with a contractor experienced in line construction and labour productivity in remote areas of British Columbia.

Unit costs of 138 kV and 230 kV lines were estimated at \$140,000/km and \$160,000/km respectively. These costs include allowances for:

- right-of-way survey and land acquisition costs of \$ 3,000/km
- clearing and access roads
- all pole line materials
- foundation and installation work
- camp costs
- engineering and site inspection
- contingency allowance of 15%.

Wooden pole H-frame construction has been assumed generally for both 138 and 230 kV lines. The poles would be buried direct in the ground with compacted backfill. Certain sections may require steel towers depending upon conditions.

435 MISCELLANEOUS

No consideration has been given to possible effects on routing due to Indian Reserves along the Stikine River or to Indian land claims.

No consideration has been given to possible conflict with routes which may be planned by BCHPA for the future. Such conflict could affect routing in areas of minimal land availability at reasonable cost, if such exist.

No allowance has been made for winter construction; it has been assumed that all line construction would be performed in summer.

A higher unit cost allowance has been made for System 8, based upon discussions with BCHPA as to type of terrain and effect of possible land reserves for hydro-electric development in the future at the More Creek site.

4-3



TABLE 4.1

CAPITAL COST ESTIMATE

TRANSMISSION SYSTEM 1 - 138 kV

(\$000 Cdn - 1982)

í

Mount K	Clappan to Red Chris Substation	\$
-	138 kV transmission line	23,700
-	25 kV line to mine	1,000
-	25 kV line to Eddontenajon	100
Substati	ion, allow	2,700
Red Chr	ris to Kutcho Creek area	
-	138 kV transmission line	18,600
-	69 kV line to Eaglehead	2,600
-	69 kV line to Letain	1,300
-	69 kV line to Kutcho	3,500
 25 Kv line to townsite 		300
Substati	on, allow	3,200
тс	DTAL	57.000

()



TABLE 4.2 CAPITAL COST ESTIMATE TRANSMISSION SYSTEM 2 - 138 kV (\$000 Cdn - 1982) Mount Klappan to Red Chris Substation \$ 138 kV transmission line 23,500 _ 25 kV line to mine ----1,000 -25 kV line to Eddontenajon 100 Substation, allow 2,900 Red Chris to Dease Lake area 138 kV transmission line 18,600 -25 kV line to Dease Lake _ 200 Substation, allow 3,100 Dease Lake to Kutcho Creek area 138 kV transmission line -12,600 69 kV line to Eaglehead -2,600 69 kV line to Letain 1,300 ----69 kV line to Kutcho Creek 3,600 -25 kV line to townsite 300 Substation, allow 3,200 TOTAL 73,000



TABLE 4.3

CAPITAL COST ESTIMATE - 1982 \$000's

TRANSMISSION SYSTEM 3 - 138 kV

6		\$
System 4	138 kV serving Red Chris, Kutcho Creek area. Ref. Table 4.2	73,000
Dease La	ake to Cassiar	
_	138 kV transmission line	15,300
-	69 kV to mine	400
-	69 kV to town	300
Substatio	on, allow	2,900
Cassiar 1	to Atlin (Adanac)	
-	138 kV transmission line	39,000
-	69 kV to mine	1,100
-	25 kV to town	500
Substatio	on, allow	2,500
то	TAL	\$135,000

1

()

 \bigcirc



TABLE 4.4

· · · - - -

t

CAPITAL COST ESTIMATE

 \bigcirc

 \bigcirc

()

TRANSMISSION SYSTEM 4 - 138 kV

(\$000 Cdn - 1982)

Mount Klappan to Sustut Mine	\$
 138 kV transmission line 25 kV line to town/mine 	27,800 200
Substation, allow	2,000
TOTAL	30,000

PART 4

 \bigcirc

()

HIGH VOLTAGE TRANSMISSION SYSTEMS

PART 4 - HIGH VOLTAGE TRANSMISSION SYSTEMS

Page

CONTENTS

 \bigcirc

()

 \bigcirc

4.1	INTRODUCTION	4-1	
4.2	SYSTEM ALTERNATIVES		
4.3	CAPITAL COST ESTIMATES	11 2	
		4-2	
	431 Basis of Estimates	4-2	
	432 Route Selection	4-2	
	433 Line Length	4-2	
	434 Line Costs	4-3	
	435 Miscellaneous	4-3	
4.4	OPERATING COST ESTIMATES	4_4	
4.5	SYSTEM CHARACTERISTICS	4-5	



TABLE 4.5

CAPITAL COST ESTIMATE

TRANSMISSION SYSTEM 5 - 230 kV

(\$000 Cdn - 1982)

Mount Klapp	Mount Klappan to Red Chris Substation			
- 23 - 25 - 25	0 kV transmission line kV line to mine kV line to Eddontenajon	25,300 1,000 100		
Substation, a	llow	3,100		
Red Chris to Telegraph Creek				
- 23 - 25	0 kV transmission line kV to Telegraph Creek	18,000 100		
Substation, a	llow(a)	3,700		
Telegraph C	reek to Stikine Copper			
- 23 - Su	0 kV line bstation to 69 kV say, allow(b)	18,000 2,000		
Telegraph Ci	reek to Schaft Creek			
- 23 - Su	0 kV line bstation, to 69 kV say, allow	12,700 2,000		
Townsite. A	ssume by mines			
TOTAL		_85,000		

Note: (a) Cost of substation at Telegraph Creek if town of Telegraph Creek not served from 230 kV line would be about \$2.5 million.

(b) Substation voltage would depend upon mine requirements and distribution systems proposed.



TABLE 4.6

CAPITAL COST ESTIMATE

C

 \bigcirc

 \bigcirc

TRANSMISSION SYSTEM 6 - 230 kV/138 kV

(\$000 Cdn - 1982)

System 5 - 230 kV serving Red Chris,	\$
Ref. Table 4.5.	85,000
Major switching substation on Stikine River for 230 kV/138 kV switching and transforming, located on 230 kV line about half way between Red Chris and Telegraph Creek.	10,000
Stikine substation to Dease Lake	
 138 kV transmission line 25 kV line to Dease Lake 	11,000 200
Substation at Dease Lake, allow	3,100
Dease Lake to Eaglehead/Letain/ Kutcho Creek, 138 kV, See Table 4.2.	23,700
TOTAL	133,000




TABLE 4.8

CAPITAL COST ESTIMATE

TRANSMISSION SYSTEM 8 - 230 kV

(\$000 Cdn - 1982)

Mount Klappan to Bob Quinn area(a)	\$
 230 kV transmission line 25 kV to camp site (b) 	18,000 Not incl.
Substation at Bob Quinn area, allow	2,800
Bob Quinn to switching station	
- 230 kV line	20,200
Switching station, allow	2,000
Switching station to Stikine Copper	
 230 kV line 69 kV line to mine, say substation to 69 kV, say 	26,500 300 1,700
Switching Station to Schaft Creek	
 230 kV line 69 kV line to mine, say substation to 69 kV, say 	11,000 500 2,000
Townsite supply. Assume by mines	
TOTAL	85,000

Note:

(a) There is doubt as to the practicability of this route.

- (b) Load at Bob Quinn Camp assumed much too small for cost of transforming from 230 kV to 25 kV say.
- (c) Substation voltage would depend upon mine requirements and distribution systems proposed.

4.4 OPERATING COST ESTIMATES

The operating cost estimates given in Table 4.9 have been based on a percentage of the capital costs of each system. The sub-stations are assumed to be un-manned. Crews for line-work maintenance and sub-station duties are assumed to be located at reasonably central points for each system.

TABLE 4.9

TRANSMISSION SYSTEMS

OPERATING COST ESTIMATES

(\$000 Cdn - 1982)

		\$
System		Annual Operating Cost
System 1 -	Red Chris, Eaglehead Letain, Kutcho Creek	1,500
System 2 -	Red Chris, Dease Lake, Eaglehead, Letain, Kutcho Creek	1,900
System 3 -	System 2 plus Cassiar, Atlin (Adanac)	3,800
System 4 -	Sustut	800
System 5 –	Red Chris, Telegraph Creek, Stikine Copper, Schaft Creek	2,400
System 6 -	Red Chris, Eaglehead, Letain, Kutcho Creek, Stikine Copper, Schaft Creek	3,900
System 7 -	All mines	6,800
System 8 -	Stikine Copper, Schaft Creek alternative	3,500

As noted in Part 1, the power losses in the transmission system have been assumed to be included in the estimated mine loads.

4-4

i

4.5 SYSTEM CHARACTERISTICS

The power carried by a system varies as the square of the voltage. The electrical stability of the system can be significantly improved by using higher voltages. These two major considerations must be studied in any system design with accompanying economic comparisons made of the additional costs of lines and substations.

One primary consideration in the selection of line voltage for the systems chosen has been the estimation of the potential capacity of the line for a voltage drop of from 10 to 15%. For a given conductor size, the relationship between voltage drop, as it may affect the capacity of a given system, and distance to the load source is not linear, and for shorter distances with higher loads may not be the governing technical design consideration.

A second consideration arises from the nature of mining operations in which large swings in short term power demand occur (such as from start-up of large motors and, more particularly, from operation of large electric shovels). These large demand swings of short term duration have considerable influence on the fault levels for which a system must be designed. The effect of these voltage fluctuations (flicker effect) due to these swings can be reduced substantially by increasing the primary transmission voltage.

Other matters which affect the voltage of the line chosen and conductor size include:

- load characteristics of the mine(s),
- location of the mine in the systems and the effects on other customers of the mine operations,
- design of the electrical system at each mine as to type and size of large motors, and the use of compensating equipment, reduced voltage starters, and so on to minimize instability in the system,
- ice and wind loadings as they affect strength requirements of the cable for economical spans between towers.

4-5



I

(

 \bigcirc

()

DIESEL-POWERED ELECTRIC GENERATING STATIONS

PART 5 - DIESEL-POWERED ELECTRIC GENERATING STATIONS

	COr	VIEN	15	<u>Page</u>
5.1	INT	RODU	JCTION	5-1
5.2	CAE	PITAL	. COSTS	5-1
	521	Iten	ns Included	5-1
		.1	Diesel - Generator Units	5-1
		.2	Auxiliaries	5-2
		.3	Site	5-2
		.4	Buildings	5-2
		.5	Waste Heat, Recovery Portion	5-2
		.6	Commissioning	5-2
		.7	Bulk Storage Facilities	5-2
		.8	Water Supply	5-2
		.9	Housing	5-3
		.10	Spare Parts	5-3
	522	Iten	ns Not Included	5-3
		-1	Transmission Lines and Mine Substation	5-3
		.2	Waste Heat Recovery, Remainder	5-3
		.3	Access Road	5-3
5.3	OPE	RATI	NG COSTS	
	531	Unit	Costs	
		.1	Oil	5-4
		.2	Operating and Maintenance Costs	5-4
		.3	Interest and Depreciation	5-4
5.4	TYP	E OF	OIL	5-5

Table

(

- 5.1 Capital Costs, Summary
- 5.2 Operating Costs Per Annum, Summary

ĩ

5.1 INTRODUCTION

Capital and operating cost estimates for three sizes of diesel-powered electric generating plants are presented in this Part. Estimates are provided for installed capacities of 12.5 MW, 25 MW, and 50 MW respectively. These order-of-magnitude estimates may be used to compare costs of diesel-powered electric generation at selected potential mine locations with the costs of generation and transmission of power from Mount Klappan. They may be used also to compare coalfired power generation costs at Mount Klappan at the lower station capacities considered. Capital cost estimates are given in Table 5.1 and operating cost estimates in Table 5.2.

5.2 CAPITAL COSTS

521 ITEMS INCLUDED

General descriptions of the plants follow, together with some comments on the basis of the choice of sizes as they may affect the interpretation of the cost estimates given in this Section.

.1 Diesel-Generator Units

The numbers of units and installed capacities are representative of typical installations. The capacities stated are based upon continuous ratings of the diesel units, however, for the purposes of establishing firm capacity for each plant, at least one unit should be deducted from the total installed capacity to allow for downtime and major overhauls. For example:

- a 12.5 MW Plant could consist of five, 6-cylinder units
 (d 2,500 kW output each; the firm capacity would be a maximum of 10 MW;
- a 25 MW Plant could consist of six, 9-cylinder units @
 4,200 kW output each; the firm capacity would be a maximum of 21 MW;
- a 50 MW Plant could consist of seven, 16 cylinder units, (V-configuration) (d 7,500 kW output each; the firm capacity would be a maximum of 42.5 MW.

The diesel units would be the low-speed type designed for reliability and long life. For the purposes of the cost estimates, the use of No. 2 diesel oil has been assumed which would not require special heating or filtration treatment, although No. 1 oil may be necessary in mid-winter. Further 5-1 discussion on the influence of the type of diesel oil is contained in Section 5.4.

The actual choice of size of the spare unit (or units) is dependent upon detailed study of the load characteristics of each installation, and consideration of such matters as spare parts, peak load variations and overall efficiencies. For example, the choice of two, 8-cylinder units rather than one, 16-cylinder as the spare unit(s) may prove better on detailed analysis for a particular installation.

.2 Auxiliaries

Auxiliaries would include an overhead crane, compressor systems, inside oil storage and pump facilities, heat exchangers, intake and exhaust systems, and so on.

.3 Site

A level site with no unusual foundation conditions has been assumed.

.4 Building

An insulated, metal-clad building designed for a bridge crane has been assumed, with concrete foundations on spread footings.

.5 Waste Heat, Recovery Portion

Recovery of a portion of the waste heat sufficient for operation of the plant within the building has been assumed.

.6 Commissioning

An allowance has been made for commissioning to be done by the manufacturer.

.7 Bulk Storage Facilities

The capacity of bulk storage tanks will depend upon the effects of winter, spring break-up conditions, and other such matters on deliveries of oil. An approximate allowance for about 30 days storage has been included, including cost of the oil to fill the tanks.

.8 Water Supply

An allowance has been made for the supply of raw water for cooling, makeup, and station requirements. Actual costs can vary considerably depending upon relative locations and reliability of the water supply.

5-2

.9 Housing

An allowance has been included for the cost of providing housing for the Superintendent and General Foreman. No attempt has been made to allow for possible subsidy plans or camp allowances for operators which can vary widely with specific site conditions.

.10 Spare Parts

An allowance has been included for an initial supply of spare parts.

522 ITEMS NOT INCLUDED

A listing of the main items which have not been included in the estimates is given in this section, together with a brief discussion on each item.

.1 Transmission Lines and Mine Substation

The transmission line voltage and length from the diesel plant are site-specific for any installation, as are the substation requirements at the point of delivery.

.2 Waste Heat Recovery, Remainder

Use of waste heat from the exhaust gases and cooling water for such purposes as space heating elsewhere and for ore drying can be economical depending upon the relative location of these facilities to the plant. Recovery of such heat can cause the overall plant efficiency to increase to above 90 percent.

.3 Access Roads

Access to the site and maintenance of the roads is not included in the estimate.

5.3 OPERATING COSTS

- 531 UNIT COSTS
 - .1 Oil

The fuel oil has been assumed to be normal diesel fuel, (No. 2) for this estimate.

The price of oil delivered, including provincial sales tax, has been assumed as \$0.37 per litre, which reflects current cost experience in the area.

.2 Operating and Maintenance Costs

The operating and maintenance costs have been estimated based on ratios obtained from an existing installation.

.3 Interest and Depreciation

No allowance has been made for the cost of interest or depreciation.

5.4 TYPE OF OIL

Generally speaking, heavier (more viscous) oils are cheaper than the lighter oils and have a higher calorific value per litre. The diesel engines can be readily designed to accept heavy oils, however, the advantages of the cost of heavy oil must be weighed against the increased cost of handling, transportation, storage and treatment (before use) which is necessary when using these oils.

The use of the very light No. 1 Diesel oil in winter should present few problems, and, similarly use of No. 2 oil (commonly called "normal" diesel) in summer is very satisfactory from an operational viewpoint. As heavier oils are used, measures must be taken in colder weather to use heated barges, trucks and storage tanks and insulated pipe systems so that the oil is sufficiently fluid to handle. Further, in the diesel-powered electric generating station a treatment plant must be installed for removing sludge and tars from the heavy oil. (A typical capital cost of such treatment equipment is in the order of \$350,000 to \$400,000 for a 25 MW plant). The sludge must be stored until disposed of or used in summer for such purposes as dust control on roads. Disposal of sludge from heavy oils can present environmental problems.

The potential use of concentrate trucks for back-haul of diesel oil is also complicated by use of heavy oils for the reasons mentioned above. This should be studied, along with the volumes of light diesel fuels used by mobile equipment, for any particular location.

Use of heavy diesel may be practical in cases where the power plant is close to a primary delivery and major storage depot, where heating of large volumes of oil can be handled economically. For very large installations, the increased efficiencies obtained from larger dieselpowered electric generating units may be a significant factor in any comparison of fuel usage involving very large volumes.

The economics of use of plants located some distance from such a depot can only be established by detailed study of particular operations associated with the plant and the volumes and projected life of the plant concerned.

5-5

TABLE 5.1

CAPITAL COST ESTIMATE

DIESEL-POWERED ELECTRIC GENERATING PLANTS

(\$000 Cdn - 1982)

Installed Capacity	12.5 MW	25 MW	50 MW
Number of Units	5	6	7
	\$	\$	\$
Building Site	900	1,200	1,350
Major Equipment, Supply	7,100	14,600	30,000
Auxiliaries, Supply	200	275	425
Installation	500	7 <i>5</i> 0	1,300
Commissioning	50	75	125
Spare Parts	100	150	200
Sub-total	8,850	16,950	33,400
Bulk Storage Tanks	325	500	725
Oil to fill storage tanks	660	1,325	2,600
Raw Water Supply, Allow	300	500	850
Owner's Costs and Housing, Allow	115	125	125
TOTAL	10,250	<u>19,400</u>	<u>37,700</u>
Unit Cost per kW installed	820/kW	775/kW	755/kW

Note: Engineering and Contingency included.

TABLE 5.2

OPERATING COSTS PER ANNUM

DIESEL-POWERED ELECTRIC GENERATING PLANTS

AT 70 % CAPACITY FACTOR

(\$000 Cdn - 1982)

Installed Capacity	12.5 MW	25 M₩	50 MW
Number of Units	5	6	7
	\$	\$	\$
Oil, 70% Capacity Factor	7,950	15,900	31,000
Lubricating Oil	250	500	1,000
Operation and Maintenance	2,200	4,000	7,800
Supervision, Allow	70	100	100
TOTAL	10,520	20,500	39,900
Electrical output	76.65 MWh	153.3 MWh	306.6 MWh
Unit cost/kWh	13.7 ¢/kWh	13.4 ¢/kWh	13.0 ¢/kWh

PART 6

ENGINEERING AND CONSTRUCTION SCHEDULES

PART 6 - ENGINEERING AND CONSTRUCTION SCHEDULES

Page

CONTENTS

6.1	INTRODUCTION	6-1
6.2	COAL-FIRED POWERPLANTS 621 General	6-1 6-1
	622 50 MW to 150 MW Stations623 200 MW to 250 MW Stations	6-1 6-1
6.3	HIGH VOLTAGE TRANSMISSION LINES	6-2
6.4	DIESEL-POWERED ELECTRIC GENERATING PLANTS	6-2

6.1 INTRODUCTION

Engineering and construction schedules are briefly presented in this part for coal-fired generating stations, transmission line construction, and diesel-powered electric generating stations. Assuming that approvals are given by governing bodies, generally it can be said that:

- coal-fired stations could be designed and constructed in four to six years depending upon the size of plant and boiler delivery times;
- transmission line land acquisition, and design and construction could take from three to four years;
- diesel-powered plants could take about two to two and one-half years, depending on the size of the station.

6.2 COAL-FIRED POWERPLANTS

621 GENERAL

The major critical item in scheduling coal-fired stations is the delivery and erection of the boilers. Depending upon the size of the units and business activity of boiler manufacturers, delivery times are typically from eighteen months to thirty months. Erection can take from eighteen to twenty-four months, including ancillaries and turbine installation. Start-up and commissioning can take six months. For multiple-unit installations, it is advisable to stage start-up of the second and following units from three to six months later depending upon the load committments. Erection time can be reduced by working two shifts or by duplicating erection crews on two or more units, however, the procedure is not expected to be as cost-efficient as for single-shift erection.

622 50 MW TO 150 MW STATIONS

Engineering and construction for the 50 MW, 100 MW and 150 MW stations would normally take approximately four to five years.

623 200 MW and 250 MW STATIONS

The larger stations, 200 MW and 250 MW, would take approximately five to six years to design and build.

6-1

6.3 HIGH VOLTAGE TRANSMISSION LINES

Time required for preliminary route surveys and land acquisition can vary with the amount of time required for government approvals, and the extent of governments support for proposed developments can be expected to have significant influence on the time required. Clearing activity depends upon such matters as logging regulations required by the Forest Ministry for scheduling of commercial log recovery, watercourse protection, land stability and requirements for brush clearing and grubbing.

Assuming that engineering and procurement would proceed during these early activities, construction of the lines and substations would take from one to two years. To avoid winter work as much as possible on line construction, sections of the line could be broken into smaller contracts to ensure completion in summer months.

In summary, depending upon the extent of government support, it could be expected that the average transmission line and substation system considered in this study would take from a minimum of three to a probable four years to complete.

6.4 DIESEL-POWERED ELECTRIC GENERATING PLANTS

Engineering and construction of the diesel-powered electric generating plants would take about two years. The delivery and installation schedule for the larger stations of 40 to 50 MW capacity would depend largely upon the capacity of the manufacturing company to deliver the units in nine to twelve months. APPENDICES

CONFIGNM

GULF CANADA RESOURCES INC.

MOUNT KLAPPAN COAL PROJECT

1**982**

MINE ASSESSMENT

VOLUME 1

SUMMARY

Coal Licence Number 7118 to 7177

7381 to 7392

and

7416 to 7432 inclusive

Cassiar Land District

NTS Map Number 104 H

Latitude Between 57°11' and 57°22'N Longitude Between 128°39' and 129°05'W

BY

PHILLIPS BARRATT KAISER Engineering Ltd. Vancouver, B.C. Canada

PBK Project No. 82054

DECEMBER 1982

00110(4)



MOUNT KLAPPAN COAL PROJECT







MOUNT KLAPPAN COAL PROJECT

MINE ASSESSMENT

VOLUME 1 SUMMARY
 VOLUME 2 GEOLOGY
 VOLUME 3 MINING
 VOLUME 4 COAL PREPARATION FACILITIES
 VOLUME 5 INFRASTRUCTURE



()

(

 \bigcirc

)

٢

SUMMARY

CONCLUSIONS

Bearing in mind the preliminary nature of the geological exploration of the Mount Klappan property, the conclusions of Phillips Barratt Kaiser Engineering Ltd. are:

- 1. A resource of anthracite of commercial importance exists at the Mount Klappan project site.
- 2. Mining the property by open pit methods using conventional truck and shovel equipment appears feasible.
- 3. Washing and upgrading the coal using a jig, water-only cyclone and flotation circuit to produce a product anthracite of 20% ash appears feasible.
- 4. The laying of 85 km of track on a partially completed railway subgrade will complete the connection to Ridley Island Terminal.
- 5. With the completion of rail access to the property, unit trains of coal can be moved to the port.
- 6. The rail distance to the port can be shortened by approximately 800 km (from 1400 to 600 km) by the construction of 176 km of new line connecting the British Columbia Railway to the Canadian National Railway.
- 7. The large coal resource base, favourable stripping ratios for open pit mining and a potentially favourable rail haulage distance combine to offer the prospect of a successful mining operation.

MOUNT KLAPPAN COAL PROJECT

VOLUME 1

SUMMARY

TABLE OF CONTENTS

CONCLUSIONS

PART

- 1 EXECUTIVE SUMMARY
- 2 INTRODUCTION
- 3 PROJECT DESCRIPTION
- 4 LABOUR AND MAJOR EQUIPMENT REQUIREMENTS
- 5 INFRASTRUCTURE

MASTER TABLE OF CONTENTS

APPENDIX A

APPENDIX B



EXECUTIVE SUMMARY

 \bigcirc

 \bigcirc

PART 1 - EXECUTIVE SUMMARY

CONTENTS

Page

-

1.1	BACKGROUND	i-1

1.2 ASSESSMENT FINDINGS 1-1

1.1 BACKGROUND

The Mount Klappan property is located in northwestern British Columbia some 336 km northeast of the Ridley Island Coal Terminal now under construction. Gulf Canada Resources Inc. owns 89 whole and partial coal licenses covering some 22 000 hectares with applications pending for further licenses covering approximately 16 000 more hectares.

Preliminary geological exploration work by Gulf in 1981 and 1982 indicates an area covering less than 10% of the property to be underlain by nine anthracite seams with a combined average thickness of 21.75 metres. This area contains an inferred in situ resource of 620 million tonnes. Low sulphur clean coal products could be produced having ash contents ranging from 5% - 20%.

Phillips Barratt Kaiser Engineering Ltd. were retained in September, 1982 to carry out a mine assessment. The requirements were to develop conceptual engineering plans for two alternative mining operations and to investigate the infrastructure needed to produce and ship the product coal.

Two conceptual mine plans were developed:

- A mine to produce 5 million tonnes per year of anthracite washed to a 20% ash level. The coal preparation plant has a nominal yield of 70%. The product stripping ratio is 8.8 m³/t of product coal.
- An alternative mine to produce and ship 1 million tonnes per year of run-of-mine coal. The product stripping ratio is 7.4 m³/t.

1.2 ASSESSMENT FINDINGS

Phillips Barratt Kaiser's assessment of the mining and coal processing requirements of the Mount Klappan coal property, as presently defined, shows that open pit mining appears technically feasible and coal washing requirements can be met using conventional equipment.

In both cases, open pit bench mining was proposed using conventional truck/shovel/equipment. Waste will be transferred to external dumps by overland conveyor; dumps will be constructed by slewing stackers traversing transfer conveyors.

The coal preparation plant circuit consists of jig, water-only cyclone and flotation washing equipment.

Transportation to the Ridley Island port could begin following the laying of 85 km of track on the partially completed subgrade already extending to the minesite along the British Columbia Railway (BCR) right-of-way. This line connects with the Canadian National Railway (CNR) at Prince George. The total distance from the mine to the port is approximately 1 400 km. Construction of a new section of railway line to connect the BCR to the CNR near Hazelton would reduce the distance to approximately 600 km.

The new coal port at Ridley Island, scheduled for completion in 1983, will have a planned capacity of 12 million tonnes per year, approximately 4 million tonnes per year more than the tonnage proposed from the two coal projects now under construction in northeastern British Columbia. Design of the present Ridley Island Terminal allows for an expansion to handle 15 to 16 million tonnes per year.

PART 2

INTRODUCTION

i

()

PART 2 - INTRODUCTION

CONTENTS

(

 \bigcirc

6

2.1	BACKGROUND	2-1
	211 Coal Licences	2-1
	212 Study Assignment	2-1
	213 Subsequent Studies	2-1
	214 Quality of Product Coals	2-1
2.2	CONSULTANTS EMPLOYED	2-2
2.3	BASIC DATA PROVIDED BY	2-2
	GULF CANADA RESOURCES INC.	
24	SITE DESCRIPTION	
∠. 4	SHE DESCRIPTION	2-2
2.5	ACKNOWLEDGENENTS	~ ~ ~
2.)	ACKNOWLEDGEMEN 15	2-3

2.1 BACKGROUND

The coal occurrences of the Bowser Basin have attracted interest for more than one hundred years. However, only in the last few years has serious attention been directed to the search for anthracite in the northern portion of the basin.

Since 1979 Gulf Canada Resources Inc. has been engaged in exploration of the northern portion of the Bowser Basin. This work resulted in the acquisition of the Mount Klappan coal property in 1981.

211 COAL LICENCES

As a result of the 1981 programme Gulf Canada Resources Inc. now owns 89 whole and partial British Columbia coal licences. These licences cover a total of 22 371 hectares of land.

Following the 1982 geological field programme, an application was filed for 53 coal licences covering 15 901 hectares.

Gulf Canada Resources Inc. has also reapplied for approximately 1 000 hectares or land bordering the Spatsizi Plateau Wilderness Park. This additional area brings the total area either currently held by licences or under application to 38 272 hectares.

212 STUDY ASSIGNMENT

In September 1982, Gulf Canada Resources Inc. retained Phillips Barratt Kaiser Engineering Ltd. to carry out a mine assessment. PBK was retained to prepare a conceptual engineering study for the mining and washing of anthracite coal and to investigate the regional infrastructure required. Examination of two levels of production was called for:

- 5 million tonnes per year of product shipped as anthracite washed to a 20% ash content; and,
- I million tonnes per year of product shipped as run of mine anthracite.

213 SUBSEQUENT STUDIES

In December 1982, Phillips Barratt Kaiser Engineering Ltd. was asked to perform additional studies to assess the feasibility of mining and washing anthracite coal to obtain product coals with ash contents at the 5% and the 10% levels.

2-1

214 QUALITY OF PRODUCT COALS

The 5 million tonnes per year of anthracite washed to a 20% ash content and the 1 million tonnes per year of run-of-mine anthracite are intended for use as briquetting coal.

The 5% and 10% ash level coals, which are the subject of subsequent studies, would be considered for purposes other than as briquetting coal.

2.2 CONSULTANTS EMPLOYED

Phillips Barratt Kaiser Engineering Ltd. (PBK) has acted as the prime consultant for this study and has evaluated the mining, coal preparation and the regional infrastructure requirements of rail and road access.

Other services have been provided by the following consultants:

- Environmental Management Associates environmental assessment.
- Price Waterhouse Associates Ltd. socio-economic assessment.

2.3 BASIC DATA PROVIDED BY GULF CANADA RESOURCES INC.

The basic geological data, geological cross sections and coal quality data were supplied by Gulf Canada Resources Inc.

2.4 SITE DESCRIPTION

The Mount Klappan coal project site at latitude 57° 15' N and 128° 50' W lies some 336 km. northeast of Prince Rupert in northern British Columbia. A British Columbia Railway (BCR) right-of-way connecting to Prince George runs through the project site. At present, the rail track laid from Prince George stops 85 km to the south of the Mount Klappan project site.

The topography is characterized by broad, open subalpine valleys and generally subdued mountains with elevations ranging from 1100 to 2000 m. The licence area is located in the headwaters of the Little Klappan and Spatsizi Rivers.

Present site access is by air to a 1000 m airstrip on the property. Road access could be established from Highway 37 (Stewart-Cassiar) via the Ealue Lake road and then the BCR subgrade south to the project site. Bridges for one river crossing and two stream crossings would have to be completed.

2.5 ACKNOWLEDGEMENTS

We wish to acknowledge the assistance, encouragement and support of the Coal Department staff of Gulf Canada Resources Inc. in this assessment.

PART 3

Ç

PROJECT DESCRIPTION

PART 3 - PROJECT DESCRIPTION

CONTENTS

1

()

 \bigcirc

()

			Page
3.1	INTR	RODUCTION	3-1
3.2	GEO	LOGY AND COAL QUALITY	
	321	Field Programmes	3-1
	322	General Geology	3-1
		.1 Geological Setting	3-1
		.2 Resources	3-2
		.21 Quantity	3-2
		.22 Quality	3-2
	323	Mining Area Geology	3-3
		.1 Description	3-3
		.2 Resources	3-3
		.21 Quantity	3-3
		.22 Quality	3-4
3.3	5 Mt	tpy CASE	
	331	Mining	3-4
	332	Coal Preparation	3-5
	333	Administration	3-6
3.4	1 Mt	tpy CASE	
	341	Mining	3-6
	342	Coal Handling	3-7
	343	Administration	3-8
3.5	REC	CLAMATION	3-8
Table

- 3.1 Material Movement Schedule 5 Mtpy Case
- 3.2 Material Movement Schedule 1 Mtpy Case

Figure

- 3.1 Simplified Preparation Plant Flow Sheet
- 3.2 Organization Chart 5 Mtpy Case
- 3.3 Oraganization Chart 1 Mtpy Case

Plate

- 3.1 Hobbit-Broatch Resource Area -General Arrangement - 5 Mtpy
- 3.2 Hobbit-Broatch Resource Area -General Arrangement - 1 Mtpy

3.1 INTRODUCTION

Two alternative mining concepts were called for:

- 5 million tonnes per year (5 Mtpy Case) of anthracite, washed to a 20% ash level and then shipped as the product; and
- I million tonnes per year (1 Mtpy Case) of run-of-mine anthracite shipped as the product.

The mining assessment performed by Phillips Barratt Kaiser Engineering Ltd. is based on the geological data and geological cross sections provided by Gulf Canada Resources Inc. This data was obtained from field programmes conducted in 1981 and 1982.

3.2 GEOLOGY AND COAL QUALITY

321 FIELD PROGRAMMES

The following field programmes were conducted:

1981 Programme

An initial geological assessment consisting of mapping, trenching and surface sample collection was made in the late summer and early fall of 1981. Data gathered from this work guided the design of the 1982 exploration programme.

1982 Programme

Based on the 1981 results, a programme of geological mapping, trenching and diamond drilling was conducted during the summer of 1982. 50 hand trenches, with an aggregate length of over 285 metres, were dug in coal exposures and 7 diamond drill holes were drilled for a total of 1223 metres. Coal samples taken during the coring programme were subjected to detailed analytical testing and washability studies.

322 GENERAL GEOLOGY

.1 Geological Setting:

The Mount Klappan coal property is underlain by the coal bearing Upper Jurassic to Lower Cretaceous Klappan Sequence. Twelve seams with a cumulative average thickness of 25.2 metres exist within the 300 to 350 metre interval of the Middle Klappan Sequence.

.2 Resources

.21 Quantity

The property is estimated to have a resource potential of 3 billion tonnes of coal of which 890 million tonnes is calculated to be in the inferred resource category. This inferred category comprises:

Resource Area	Million <u>Tonnes</u>
Hobbit-Broatch Lost-Fox Summit	620 240 _30
Total	890

The Hobbit-Broatch Resource Area is the portion of the property which was evaluated in this assessment. Thirty-four trenches have been excavated in coal outcrops, and 5 diamond drill holes have been completed in the area.

Coal seams A to K were intersected by the drill holes. Seams C and D were excluded from the resource calculations because they have weighted average thicknesses less than 0.5 metres. The weighted average thickness of the 9 seams used in the resource calculation is 21.75 metres.

.22 Quality

The raw coal analysis for the Hobbit-Broatch Resource Area, on an air dried basis is:

Proximate Analysis		
Residual Moisture	1.42	%
Ash	30.87	%
Volatile Matter	8.36	%
Fixed Carbon	59.35	%
Total Sulphur	0.88	%
Thermal Value	5238	kcal/kg
	2.19	MJ/kg
Hardgrove Grindability Index	54	0

323 MINING AREA GEOLOGY

.1 Description

Four of the five Hobbit-Broatch Resource Area diamond drill holes influence the mining area selected. These are DDH-82001, 82002, 82003 and 82004.

Only seams E through K were included in this evaluation because seams A through D are below the depth of the diamond drill holes which influence the mining area selected.

.2 Resources

.21 Quantity

For production scheduling and mine planning purposes, a block model approach was adopted to determine the quantities and qualities of coal available in the mining area selected. A standard grid was established based on the geological cross sections available. The grid dimensions were:

- vertical (indexed to the cross section elevations) -50 metres;
- width (along the cross section axis) 200 metres;
- length (perpendicular to the cross section 250 metres on each side) 500 metres.

The volume of coal in each block was estimated by actual measurement of the width dimension on the cross sections and multiplication of this number by the block length of 500 metres and by the seam thickness.

The total estimated quantity of run of mine coal required for the life of the mine used in this assessment is 146 550 000 tonnes in the 5 Mtpy Case.

.22 Quality

The composite raw coal quality analysis (calculated by PBK) on an air dried basis for the mine area selected is:

Residual Moisture	1.50	%
Ash	29.85	%
Volatile Matter	6.82	%
Fixed Carbon	61.83	%
Total Sulphur	0.74	%
Thermal Value	22.26	MJ/kg
	5325	kcal/kg

3.3 5 Mtpy CASE

331 MINING

The development of a conceptual mine plan has followed the approach of selecting the low stripping ratio areas for the 5 million tonne per year level of production.

The mining concept has not been optimized, however, engineering judgement has been exercised to incorporate improvements where appropriate.

The mining system is an open pit benching approach with waste disposal to waste dumps. Transfer of waste from the mine to the waste dump is by trucks to a hopper and then by surface conveyor to the dump. Coal transportation to the coal preparation plant is by truck.

Plate 3.1 presents the general arrangement of the 5 Mtpy case minesite and illustrates the mining and dump areas as well as mine limits.

The conceptual design is an open pit benching operation with the box cut sited in the southern part of the Hobbit-Broatch Area. The mining progresses in a northerly direction, with benches running east-west.

Key to developing a workable concept was the handling of the mine waste rock. A waste dump is necessary for the nine years required to complete the box cut. By Year 9, some capacity is available for waste backfilling, however, this capacity is limited because the pit bottom is rising, and the use of the external waste dumps will be necessary until Year 15. At this point, there is sufficient space in the mined out area to dispose of all mine waste by backfilling. Also, the material balance is such that this can be accomplished without purchasing additional mine trucks.

The waste rock (which has been blasted) is hauled by end dump trucks to a conveyor hopper sited close to the mine crest.

A 2 134 mm belt elevates the rock to the mountainside waste dump. Rock passes over a transfer belt to a lateral conveyor which runs on top of the dump. The lateral conveyors are traversed by stackers with 50 m slewing booms. A system of lateral conveyor extension and radial slewing is used to build the dump. Two waste dump wings provide for continuity of disposal.

Over the life of the mine, a total of $944 \times 10^6 \text{ m}^3$ of waste has to be removed. Up to Year 15, the waste conveyors handle 538×10^6 m³. The remaining 406 x 10^6 m^3 is backfilled into the mined-out area. Table 3.1 presents the material movement schedule for the 5 Mtpy case.

Conventional mining equipment such as drills, shovels and end dump trucks have been used as the basis for the capital and operating cost estimates.

A total of 106 x 10⁶ t of product coal is shipped, to give a product stripping ratio of 8.8 (m³/t). Run-of-mine (R.O.M) coal delivered to the preparation facilities is 147 x10⁶ t for an R.O.M. stripping ratio of 6.3 (m³/t).

332 COAL PREPARATION FACILITIES

The estimated product quality and preliminary flowsheet design presented are based on producing the following:

- 5,000,000 tonnes of product coal per year having:
 - an ash content of 20% (dry basis) and
 - a maximum total moisture (residual + surface moisture) of 8%.

Several preparation plant processing circuits were analyzed to assess the clean coal recovery options. A design consisting of jig, water only cyclones and flotation equipment was selected to meet the product criteria specified. Figure 3.1 presents a simplified preparation plant flow sheet. The preparation facility consists of:

- a raw coal handling system including a truck dump, rotary breaker and a 10 000 tonne raw coal silo;
- a 1 100 tph preparation plant;
- two fluid bed thermal dryers each capable of a 25 tph moisture evaporation rate;
- clean coal handling including two 10 000 tonne capacity clean coal silos;
- ground storage and a unit train loading system.

The proposed concept incorporates industry-proven circuitry for efficient operation and minimum down time.

The following product coal specification will be met:

		Run-of-Mine*		Pro	duct
		Air	As	Air	As
		Dried	Rec'd	Dried	Rec'd
		Basis	Basis	Basis	Basis
Moisture	%	1.5	5.0	1.5	7.50
Ash	%	32.20	31.07	19.70	18.50
Volatile Matter	%	6,58	6.35	7.80	7.33
Fixed Carbon	%	59.70	57.58	71.00	66.67
Sulphur	%	0.74	0.71	0.74	0.69
Thermal value	MJ/kg	21.36	20.61	26.74	25.11
	kcal/kg	5116 4	933	6398	6008
Yield		-	-	70.0	71.9

* Includes out-of-seam dilution

333 ADMINISTRATION

Appropriate minesite facilities will be provided. An organization chart is presented in Figure 3.2.

3.4 1 Mtpy CASE

341 MINING

The development of a conceptual mine plan for the run-of-mine coal to be shipped as product has followed the approach of selecting the low stripping ratio areas. The 1 Mtpy Case low stripping ratio area was restricted to the portion of the Hobbit-Broatch Resource Area totally within the influence of drill hole DDH 82-001. Drill core data indicates this area to have the lowest in-seam ash levels.

The mining concept has not been optimized, however, engineering judgement has been exercised to incorporate improvements, where appropriate.

The mining system is an open pit benching approach with waste disposal to waste dumps. Transfer of waste from the mine to the waste dump is by trucks to a hopper and then by surface conveyor to the dump. Coal transportation to the coal preparation plant is by truck.

Plate 3.2 presents the general arrangement of the mine site for the 1 Mtpy Case. The pit crest limit is shown and also the size of the waste dump after 20 years of operation.

As in the 5 Mtpy Case, the concept developed is an open pit benching operation. Mining progresses in a northerly direction from a box cut and the benches run east-west. Blasted waste rock is hauled out of the pit by truck to a conveyor hopper. A 1 066 mm conveyor belt transfers and elevates the waste to the mountainside. The same concept of waste disposal is employed as in the 5 Mtpy Case. All mine waste (148 x 10^6 m³) must be transferred to the waste dumps because seams of coal remain below the bottom of the pit. Table 3.2 presents the material movement schedule for the 1 Mtpy Case.

Conventional mining equipment (as is in the 5 Mtpy Case), was used in developing the concepts and forms the basis for cost estimating.

A total of 20 x 10^6 t of anthracite coal is shipped. The product stripping ratio is 7.4 (m³/t).

342 COAL HANDLING

No coal preparation facilities are called for in this mining case. Run-of-mine coal will be transferred directly to the coal silos after passing through a rotary breaker. The coal handling system incorporates:

- truck drump and rotary breaker station;
- a 10 000 tonne raw coal silo;
- ground storage and a unit train loadout system.

The estimated quality of coal shipped (R.O.M. coal, no dilution) is:

(This analysis is from DDH 82-001, which totally influences the 1 Mtpy Case mining area).

		Air Dried <u>Basis</u>	As Rec'd <u>Basis</u>
Moisture	%	1.5	5.0
Ash	%	27.3	31.07
Volatile Matter	%	7.2	6.8
Fixed Carbon	%	65.1	61.8
Sulphur	%	0.75	0.71
Thermal value	MJ/kg	24.7	23.5
	kcal/kg	5904	56 18

343 ADMINISTRATION

For this fly-in case appropriate facilities will be provided at the minesite. Some staff will work in an office in Smithers. The organization chart in Figure 3.3 records the work locations.

3.5 RECLAMATION

For this assessment PBK has adopted a conventional reclamation approach.

Topsoil and root zone material will be stripped and stockpiled for future use in reclamation. Once the mine cycle has progressed sufficiently, soil spreading can be carried out to facilitate revegetation on waste dumps, roads, etc. During later years in the mine life, topsoil can be stripped and taken directly to the reclamation area. 5 Mtpy CASE

·			··· · · · · · · · · · · · · · · · · ·										- · · · · · · · · · · · · · · · · · · ·		
	Units x 10 ³	-4	-3	-2	-1	ł	2	YEARS 3	٠	5	6- 10	11-15	16-20	21-23	Total
Product											·····				
Shipped	t	-	-	-	500	3 000	5000	5 000	5 000	5 000	25 000	25 000	25 000	7 379	105 879
Mine															
Waste Mine	m ³	-	-	-	3 030	20 382	39 471	38 508	38 270	40 806	237 423	229 630	215 675	66 575	929 770
Coal ROM	t	-	-	-	692	4 152	6 921	6 921	6 921	6 921	34 603	34 603	34 603	10 213	146 550
Statistics Strip Ratio (ROM)	m ³ /t	-	-	-	4.4	4.9	5.7	5.6	5.5	5.9	6.9	6.6	6.2	6.5	6.3
Mine Construction															
Roads Ditches Pond Topsoil	m ³ m3 m3	- -	160 - -	139 115 294	11 - -	11 - -	11 	11 - -	11 - -	11 - -	101 - -	114 - -	114 - -	-	69803 115 294
Waste Disposition															
Waste from Prep Pit	m3	-	-	-	67	402	671	671	671	671	3 356	3 356	3 356	989	14 211
Waste to Dump Waste to Pit	m ³ m ³	-	-	-	3 097	20 784	40 142 -	39 179 -	38 941 -	41 477 -	240 779 -	113 903 119 083	219 031	67 564	538 303 405 678
Reclamation													I		
Topsoil Excavated Mine Area Waste Dump Area Topsoil Rehandled Topsoil Spread	m ³ m ³ m ³	- - -	- - -	352 302 -	352 302	352 302 -	352 302 -	352 302 -	352 176 362 538	352 176 362 538	503. 880 1 813 2 693	503. 528 935 1 967	403 - 478 882	529 529	3 878 3 270 4 481 7 148

All volumes in bank equivalents

C

C

Decimal places in the above table have been rounded out for clarity of presentation.

* Does not apply to the stripping ratio.

MATERIAL MOVEMENT SCHEDULE - 5 Mtpy Case

С

С





. . .

.

-- - -

-- •



1 Mtpy CASE

.

										i.			ł		
							. <u></u>								
	Vnits x 10 ³	-4	-3	-2	-1	1	2	YEARS 3	٠	5	6-10	11-15	16-20	21-23	Total
Product				<u></u>								<u> </u>	1		<u> </u>
Shipped		-	-	-	-	502	1 000	1 000	1 000	1 000	5 000	5 000	5 000	508	20 010
Mine															
Waste Mine	m ³	-	-	-	-	7 533	10 555	8 421	8 176	7 871	37 228	33 512	32 046	3 405	148 747
Coal ROM	t	-	-	-	-	502	1 000	1 000	1 000	1 000	5 000	5 000	5 000	508	20 010
Statistics Strip Ratio (ROM)	m ³ /t	-	-	-	*	15	10	8.4	8.2	7.9	7.4	6.7	6.4	6.7	7.4
Mine Construction															
Roads Ditches Pond Topsoil	m3 m3 m3	- - -	-	10 - -	224 89 143	9 - -	9 - -	6 - -	6 - -	6 - -	32 - -	6.0 - -	-	- - -	331 89 143.5
Reclamation													1 		
Topsoil Excavated Mine Area Waste Dump Area Topsoil Rehandled Topsoil Spread	m ³ m3 m3 m3		- - -	- - -	454 51 - -	82 50 - -	82 49 - -	42 48 - -	42 47 -	42 46 - -	126 179 76 255	150 15 163	120 45 165	- 265 265	871 740 401 850

All volumes in bank equivalents

C

C

Decimal places in the above table have been rounded out for clarity of presentation.

Does not apply to the stripping ratio.

TABLE 3.2

MATERIAL MOVEMENT SCHEDULE - 1 Mtpy Case

- -



* These positions will be located in the Smithers office

C

- -- -- - --

** These positions will be in both Smithers office and Mine.

ORGANIZATION CHART I Mtpy Case



PART 4

()

()

 \bigcirc

LABOUR AND MAJOR EQUIPMENT REQUIREMENTS

PART 4 - LABOUR AND MAJOR EQUIPMENT REQUIREMENTS

CONTENTS

 $\tilde{()}$

4.1	INTE	INTRODUCTION						
4.2	5 Mt	tpy CASE						
	421	Major Equipment Requirements						
		.1 Mining	4-1					
	422	Labour Requirements						
		.1 Construction	4-1					
		.2 Operations	4-1					
4.3	l Mt	tpy CASE						
	431	Major Equipment Requirements						
		.1 Mining	4-1					
	432	Labour Requirements						
		.1 Construction	4-2					
		.2 Operations	4-2					

Table

- 4.1 Major Mining Equipment List 5 Mtpy Case
- 4.2 Summary of Labour 5 Mtpy Case
- 4.3 Major Mining Equipment List 1 Mtpy Case
- 4.4 Summary of Labour 1 Mtpy Case

4.1 INTRODUCTION

PBK has developed plans for mining and coal preparation which incorporate present day technology and utilize standard equipment.

4.2 5 Mtpy CASE

421 MAJOR EQUIPMENT REQUIREMENTS

.1 Mining

Table 4.1 presents a summary listing of the major units of mining equipment needed for the 5 Mtpy Case.

422 LABOUR REQUIREMENTS

.1 Construction

For this assessment, the following levels of construction labour have been estimated.

	Year -3	Year -2	Year -1	Year 1	Total Man Years
5 Mtpy	150	900	1200	250	2500

.2 Operations

Table 4.2 presents a summary of total operations labour requirements for the 5 Mtpy Case.

4.3 1 Mtpy Case

431 MAJOR EQUIPMENT REQUIREMENTS

.1 Mining

Table 4.3 presents a summary listing of the major units of mining equipment needed for the 1 Mtpy Case.

4-1

T

432 LABOUR REQUIREMENTS

.1 Construction

For this assessment, the following levels of construction labour have been estimated.

	Year -2	Year -1	Year 1	Total Man Years
1 Mtpy	250	450	175	875

.2 Operations

Table 4.4 presents a summary of total operations labour requirements for the 1 Mtpy Case.

5 Mtpy CASE

TABLE 4.1

MAJOR MINING EQUIPMENT LIST

()

5 Mtpy Case

This sample listing is for year 5.

Mining Equipment	Number Of Units	kW <u>Total</u>
Shovel - 24.5 m ³	5	4 000
Shovel - Hyd.	3	-
Truck 154 t	45	-
OB Drill BE-55R	4	640
OB Drill BE-2450	2	-
F.E.L. L-800	1	-
Dozer D9L	13	-
Dozer 834	1	-
Conveyor System		
Crusher System	1	

	-
Overland Conveyors	1
Transfer Conveyor	2
Travelling Stacker	2

TABLE 4.2 SUMMARY OF LABOUR - 5 Mtpy Case YEARS 16-20 6-10 11-15 -1 Mining Non-supervisory Supervisory Preparation

C

Non-supervisory

Supervisory

Administration

Staff

TOTAL

1 Mtpy CASE

•

15

TABLE 4.3

MAJOR MINING EQUIPMENT LIST

1 Mtpy CASE

This sample listing is for year 5.

Mining Equipment	Number of Units	kW <u>(Total)</u>
Shovel - 24.5 m ³	1	800
Shovel - hyd.	1	-
Truck - 154 t	11	-
OB Drill BE-55R	1	160
OB Drill BE-2450	1	-
FEL L-800	1	-
Dozer D8L	6	-
Dozer 834	1	-

Conveyor System

()

()

C

Crusher System	1
Overland Conveyor	1
Transfer Conveyor	2
Traveling Stacker	2

С		С					C			
		SUMMA	TAB RY OF LAI	LE 4.4 BOUR - 1 1	Mtpy Case					
	1	2	3	YEARS 4	5	6-10	11-15	16-20	21	
Mining										
Non-supervisory Supervisory	186 35	234 35	210 35	213 35	211 35	211 35	208 35	212 35	166 30	
Handling										
Non-supervisory Supervisory	20 9	20 9	20 9	20 9	20 9	20 9	20 9	20 9	20 9	
Administration										
Supervisory	59	59	59	59	59	59	59	59	31	
TOTAL	309	357	333	336	334	334	331	335	256	

PART 5

 \bigcirc

 \bigcirc

 \bigcirc

INFRASTRUCTURE

PART 5 - INFRASTRUCTURE

ŗ

1

L

Page

CONTENTS

5.1	INTRODUCTION	5-1
5.2	RAILWAY	5-1
5.3	PORT	5-1
5.4	SLURRY PIPELINE	5-2
5.5	ROAD ACCESS	52
5.6	ACCOMMODATION	5-2
5.7	ELECTRICAL POWER	5-3
5.8	WATER	5-3
5.9	SOCIO-ECONMIC IMPACT	5-3
5.10	ENVIRONMENT	5-4

Plate

 \bigcirc

1.1 Regional Access Map

5.1 INTRODUCTION

Volume 5 of this mine assessment report for the Mount Klappan project includes the regional and site infrastructure. The components of this infrastructure are railway, pipeline, road, port, electrical power, water and townsite requirements. Plate 1.1 (Regional Access Map) shows the locations of the alternative rail and road routes investigated.

The soci-economic and environmental parts of this volume were provided by consultants retained directly by Gulf Canada Resources Inc.

5.2 RAILWAY

The system considered for transportation of the coal from the minesite to the port is by rail.

The railway evaluation consists of the following items:

- the completion of the existing British Columbia Railway (BCR) line from Chipmunk to the Mount Klappan property. The distance is 85 km;
- an investigation into four possible cutoff routes (varying from 176 to 196 km) to connect the BCR line to the Canadian National Railway (CNR) line near Hazelton. Any one of these cutoffs would reduce the total haulage distance by approximately 50% of the rail distance via Prince George, which is approximately 1,400 km; and,
- the preparation of preliminary capital cost estimates for the main line completion and for three of the four alternative cutoffs investigated.

The shortest cutoff distance (176 km) along the Skeena River (Railway Route No. 2) is the preferred route. This gives a total rail haul distance of approximately 600 km to Ridley Island.

5.3 PORT

The shipping port chosen for this evaluation is the coal terminal at Ridley Island, now under construction to handle the coal which will be produced in northeastern British Columbia.

No evaluation was made as to the capability of this facility to handle the tonnage being considered for the Mount Klappan project. However, when completed in 1983, the planned capacity of this terminal will be 12 million tonnes per year, approximately 4 million tonnes per year more than the tonnage proposed from the Quintette and Bullmoose Coal Projects now under construction in the northeast. The Ridley Island Coal Terminal can be expanded to handle 15 to 16 million tonnes per year as the demand increases.

5.4 SLURRY PIPELINE

The possibility of pumping coal in slurry form from Mount Klappan to the port of Stewart has been considered as an alternative to railway transportation. Preliminary estimates put the length at some 225 km.

Construction and operation of a pipline handling fine coal should not present a problem. A detailed investigation into this option was not made, however, because the final coal markets have not been identified and, therefore, the size specifications of the shipped products is not known.

5.5 ROAD ACCESS

An investigation was made to determine the feasibility of building a road link between the existing Stewart-Cassiar Highway (Highway 37) and the Mount Klappan property. Two possible routes were studied and the lengths of the new roads required were 146 km and 226 km. Capital cost estimates were prepared for each of these alternatives.

In addition, an estimate was made for the cost of completing a temporary access road from Tatogga Lake on the Stewart-Cassiar Highway, southeast along the Little Klappan River using the existing BCR roadbed. This road would provide access during the project construction, and until a permanent access road is available.

The preferred route is the 146 km Highway Route 1, which joins Highway No. 37 at Bowser Lake.

5.6 ACCOMMODATION

Two assumptions were made for the living accommodation for the mine employees for the two mine cases considered. These are:

- the total number of personnel estimated for the 5 million tonne operation in a typical year is 1 100. It is assumed that they will be resident in a public townsite which will be constructed near the mine property.

A cost allowance per employee has been included in the capital cost estimate for assistance in developing the townsite.

the 1 million tonne per year operation is designed for a fly-in operation. The total number of on-site personnel in a typical year is 155. Employees will be air-lifted to the site from Smithers on a routine scheduled basis to cover the operation, and will live in camp accommodation while at the mine. In addition, there will be a support staff of 24 people in Smithers performing functions which do not require full-time coverage at the mine. Total personnel, in a typical year, will be 335.

5.7 ELECTRICAL POWER

This mine assessment was developed on the assumption that a source of 138 kV power will be available at the property line at current B.C. Hydro charges for power consumed.

5.8 WATER

An investigation into the availability of a source of fresh water has not been made. However, the assumption has been made that sufficient water can be obtained from the numerous water sources available in the area.

5-3

ſ

5.9 SOCIO - ECONOMIC IMPACT

The basic finding of this investigation is that no socio-economic impact was found which precludes the development of the Mount Klappan coal project. Some of the positive impacts are:

5 Mtpy Case

- up to 1 100 direct and 500 indirect new jobs in the area
- regional income increased by \$40 million per year
- total provincial employment increased by 2 750 and provincial income by \$50 million per year
- some 2 500 man years of construction labour to develop the mine and build the support facilities.

1 Mtpy Case

- 338 direct and 100 indirect new jobs created in the area
- regional income increased by some \$10 million per year
- total provincial employment increased by 676 and provincial income by some \$12 million per year
- more than 800 man-years of labour to develop the mine and construct the site support facilities.

5.10 ENVIRONMENT

The basic finding of this investigation is that no environmental factor was found which precludes development of the Mount Klappan coal project. Environmental impacts will occur as a result of project development, but proper planning, design, construction, operation and reclamation will reduce the significance of these impacts. The most significant environmental issue relates to the project's proximity to Spatsizi Plateau Wilderness Park, however, the presence of a permanent townsite would result in increased recreational use of the park, which can be viewed as a positive impact.


VOLUME 2

GEOLOGY

MOUNT KLAPPAN COAL PROJECT

VOLUME 2

GEOLOGY

TABLE OF CONTENTS

Preface

Summary

PART

1 INTRODUCTION

2 PROPERTY HISTORY

3 EXPLORATION

4 GEOLOGY

5 RESOURCES

6 COAL QUALITY

MASTER TABLE OF CONTENTS

REFERENCES

APPENDICES

PART 1 - INTRODUCTION

Page

CONTENTS

1.1	LOCATION	1-1
1.2	ACCESS	1-1
1.3	PROPERTY DESCRIPTION	1-1
1.4	OWNERSHIP	1-2
1.5	BIOPHYSICAL ENVIRONMENT	1-2

Plate

1.1	Location Map
1.2	Property Access
1.3	Licence Area
1.4	Property Geography

PART 2 - PROPERTY HISTORY

CONTENTS

<u>Page</u>

2.1 SYNOPSIS

2-1

Plate

 \bigcirc

2.1 Bowser Basin

PART 3 - EXPLORATION

CONTENTS

()

3.1	SUMMARY OF 1981 EXPLORATION PROGRAMME				
	AND RESULTS				
3.2	1982 EXPLORATION PROGRAMME				
	321	Programme Objectives and Methodology	3-1		
		.1 Objectives	3-1		
		.2 Methodology	3-2		
	322	Cartography	3-2		
	323	Logistics	3-3		
		.1 Field Camp	3-3		
.2 Mapping and Dr		.2 Mapping and Drill Support	3-3		
	324 Geological Mapping325 Hand Trenching326 Diamond Drilling		3-4		
			3-4		
			3-5		
	327	Geophysical Logging	3-5		
	328	Drill Core Logging and Sampling	3-6		
	329	Drill Core and Trench Sample Anaylsis	3-7		
	3210	Data Management	3-7		
	3211	Reclamation	3-7		
	3212	Special Projects	3-7		
		.1 Depositional Environments	3-7		
	.2 Regional Structure				

Table

3.1 Project Data Source Summary

Plate

- 3.1 1981 Speculative Resource Area
- 3.2 Camp Location
- 3.3 1982 Exploration Mapping Areas
- 3.4 Diamond Drill Holes

PART 4 - GEOLOGY

CONTENTS

(

(

()

4.1	INTR	TRODUCTION			
4.2	REGIONAL GEOLOGY				
	421	Geological Setting	4-1		
	422	Regional Stratigraphy	4-2		
		.1 Klappan-Groundhog Area Stratigraphy	4-2		
	423	Structure	4-3		
4.3	PRO	PERTY GEOLOGY	4-3		
	431	Unnamed Sequence	4_4		
	432	Klappan Sequence	4-4		
		.l Lower Klappan Unit	4-5		
		.2 Middle Klappan Unit	4-5		
		.21 Coal Seam Development	4-6		
		.3 Upper Klappan Unit	4-7		
		.4 Environment of Deposition	4-7		
	433	Malloch Sequence	4-7		
	434	Rhondda Sequence	4-8		
	435	Structure	4-8		
4.4	RES	OURCE AREA GEOLOGY	4-10		
	441	Hobbit-Broatch Resource Area	4-10		
		.1 Coal Seam Development	4-10		
		.2 Structure	4-11		
	442	Lost-Fox Resource Area	4-12		
		.1 Coal Seam Development	4-12		
		.2 Structure	4-13		
	443	Summit Resource Area	4-14		
		.1 Coal Seam Development	4-14		
		.2 Structure	4-14		

Table

- 4.1 Regional Stratigraphy
- 4.2 Table of Formations
- 4.3 Coal Seam Thickness Summary
- 4.4 Hobbit-Broatch Resource Area Seam Intersection Summary
- 4.5 Lost-Fox-Seam Resource Area Seam Intersection Summary
- 4.6 Summit Resource Area Seam Intersection Summary

Plate

- 4.1 Jurassic-Cretaceous Bowser Basin
- 4.2 Schematic Stratigraphic Column
- 4.3 Klappan-Groundhog Stratigraphy
- 4.4 Middle Klappan Unit
- 4.5 Distribution of Coal Seams
- 4.6 Schematic Geology Map
- 4.7 Schematic Cross-Section
- 4.8 Inferred Resource Areas
- 4.9 Hobbit-Broatch Resource Area Coal Seams
- 4.10 Hobbit-Broatch Correlation
- 4.11 Hobbit-Broatch Geology
- 4.12 Hobbit-Broatch Resource Area Cross-Section Summary
- 4.13 Lost-Fox Coal Seams
- 4.14 Lost-Fox-Summit Correlation
- 4.15 Lost-Fox Geology
- 4.16 Lost-Fox Resource Area Cross-Sections

PART 5 - RESOURCES

CONTENTS

 \mathbf{C}

 \bigcirc

 \bigcirc

1

ł

l

5.1	SUMMAI	5-1		
5.2	INFERR	ED RESOURCE AREAS	5-1	
	52 1	Summary	5-1	
	522	Hobbit-Broatch Resource Area	5-2	
	523	Lost-Fox Resource Area	5-2	
	524	Summit Resource Area	5-2	
	525	Low Ash Resource	5-3	
5.3	SPECUL	ATIVE RESOURCE AREA	5-3	
5.4	POTENT	TAL PROPERTY RESOURCE	5-4	
5.5	TOTAL	5-4		
5.6	PROCEI	5-4		
Tabl	e			
5.1	Summar	y of Hobbit-Broatch Resources		
5.2	Hobbit-Broatch Resource Area Weighted Seam Thickness			
5.3	Summar	y of Lost-Fox Resources		
5. 4 .	Summar	y of Summit Resources		
5.5	Low Ast	n Coal Tonnage Summary		
5.6	Speculat Resourc	tive and Potential Property e Areas-Coal Seam Thicknesses		

5.7 Coal Seam Thickness Summary

Plate

 \bigcirc

5.1 Resource Areas

- 5.2 Inferred Resource Areas
- 5.3 Speculative Resource Area
- 5.4 Potential Resource Area
- 5.5 Middle Klappan Sequence on Licences under Application

PART 6 - COAL QUALITY

CONTENTS

()

()

()

I

Pa	ge

6.1	SUMMARY				
	611	Coal Quality	6-1		
	612	Premium Coals	6-1		
	613	Briquetting Coal	6-2		
6.2	PROCEDURES AND PARAMETERS				
	621	Objectives	6-3		
	622	Methodology	6-3		
	623	Analytical Procedures	6-3		
		.1 Compositing	6-3		
		.2 Size Analysis	6-4		
		.3 Float-Sink Data	6-4		
		.4 Product Analysis	6-4		
	624	Washplant Simulation	6-4		
6.3	COA	AL RANK	6-5		
6.4	SIZE	E DISTRIBUTION	6-5		
6.5	FLO	DAT-SINK DATA	6 -6		
6.6	PRC	6-7			
	661	Low Ash Premium Coal Product	6-8		
		.1 Computed Yield	6-8		
		.2 Washplant Simulation	6-9		

661	Low	Ash P	remiu	m Coal Product (Con't)	
		.3	Prox	imate Analysis	6-10
			.31	Moisture	6-10
			.32	Ash	6-10
			.33	Volatile Matter	6-10
			.34	Fixed Carbon	6-11
		.4	Tota	l Sulphur	6-11
		.5	Calo	rific Value	6-11
		.6	Harc	grove Grindability Index	6-11
		.7	Ultir	nate Analysis	6-11
		-8	Ash	Characteristics	6-12
		.9	Mido	llings Product	6-12
662	Medium Ash Premium Coal Product			mium Coal Product	6-12
	.1	Com	puted	Yield	6-12
	.2	Wash	plant	Simulation	6-13
	.3	Prox	imate	Analysis	6-13
		.31	Mois	ture	6-13
		.32	Ash		6-13
		.33	Vola	tile Matter	6-14
		.34	Fixe	d Carbon	6-14
	.4	Tota	l Sulp	hur	6-14
	.5	Calorific Value		/alue	6-15
	.6	Hardgrove Grindability Index		Grindability Index	6-15
	.7	Ultir	nate <i>l</i>	Analysis	6-15
	.8	Ash	Chara	cteristics	6-15
	.9	Midd	llings	Product	6-15
663	Briquetting Coal Product				6-16
	.1 Introduction			n	6-16
	.2	Com	puted	Yield	6-17
	.3 Washplant Simulation			Simulation	6-17

()

()

Page

Page

663 Briquetting Coal Product (Con't)

()

 \bigcirc

.4	Proxi	imate Analysis	6-18
	.41	Moisture	6-18
	.42	Ash	6-18
	.43	Volatile Matter	6-18
	.44	Fixed Carbon	6-19
.5	Total	l Sulphur	6-19
.6	Calo	rific Value	6-19
.7	Hard	grove Grindability Index	6-19
.8	Ultin	nate Analysis	6-19
.9	Ash (Characteristics	6-20
Raw	Coal	Product	6-20
.1	Proxi	imate Analysis	6-20
	.11	Moisture	6-20
	.12	Ash	6-20
	.13	Volatile Matter	6-21
	.14	Fixed Carbon	6-21
.2	Total	l Sulphur	6-21
.3	Calo	rific Value	6-22
.4	Hard	grove Grindability Index	6-22
.5	Ultin	nate Analysis	6-22
.6	Ash (Characteristics	6-22
.7	Wash	plant	6-23
	.4 .5 .6 .7 .8 .9 Raw .1 .2 .3 .4 .5 .6 .7	.4 Proxi .41 .42 .43 .44 .5 Total .6 Calor .7 Hard .8 Ultin .9 Ash (.1 Prox .11 .12 .13 .14 .2 Total .3 Calo .4 Hard .5 Ultin .6 Ash (.7 Wash	 .4 Proximate Analysis .41 Moisture .42 Ash .43 Volatile Matter .44 Fixed Carbon 5 Total Sulphur .6 Calorific Value .7 Hardgrove Grindability Index .8 Ultimate Analysis .9 Ash Characteristics Raw Coal Product .1 Proximate Analysis .11 Moisture .12 Ash .13 Volatile Matter .14 Fixed Carbon 2 Total Sulphur .3 Calorific Value .4 Hardgrove Grindability Index .5 Ultimate Analysis .6 Ash Characteristics

Table

- 6.1 Low Ash Premium Coal Product
- 6.2 Medium Ash Premium Coal Product
- 6.3 Briquetting Coal Product
- 6.4 Low Ash Premium Coal Product
- 6.5 Medium Ash Premium Coal Product
- 6.6 Briquetting Coal Product
- 6.7 Raw Coal Product

Figure

- 6.1 Diamond Drill Core Coal Testing Programme -Part 1
- 6.2 Diamond Drill Core Coal Testing Programme -Part 2

APPENDICES

GEOLOGY

CONTENTS

APPENDIX

- A MOUNT KLAPPAN COAL PROJECT LICENCES
- B RESOURCE DATA AND CALCULATIONS
- C GEOLOGY MAPS
- D GEOLOGICAL CROSS-SECTIONS

VOLUME 3

MINING

MOUNT KLAPPAN COAL PROJECT

VOLUME 3

MINING

TABLE OF CONTENTS

PART

- 1. INTRODUCTION
- 2. GEOLOGICAL SETTING

3. MINING SYSTEMS

4. MINESITE INFRASTRUCTURE

5. ASSESSMENT PARAMETERS - MINING

MASTER TABLE OF CONTENTS

APPENDIX A

CONTENTS

()

()

(

Page

1.0 INTRODUCTION

1-1

ł

PART 2 - GEOLOGICAL SETTING

CONTENTS

				Page		
2.1	INTR	INTRODUCTION				
2.2	GEO	LOG	Y OF MINING AREA	2-1		
2.3	IN SI	2-2				
	231	2-2				
	232 Methodology			2-2		
		.1	Block Model	2-2		
		.2	Block Grid	2-2		
		.3	Quantities	2-3		
		.4	Mean Thickness of Seams	2-3		
		.5	In Situ Stripping Ratios - Mining Area	2-3		

Table

2.1 Mineable True Thickness of Coal

2.2 Mean Mineable Seam Thickness

Figure

- 2.1 Block Model Grid/Mine Grid
- 2.2 Seam Polygons
- 2.3 Composite Stripping Ratios to base of Seam E for Hobbit-Broatch Resource Area

PART 3 - MINING SYSTEMS

CONTENTS

1

- -

ł

3.1	INTR	TRODUCTION				
3.2	MINING APPROACHES					
	321	321 Overview				
	322	Top Soil	3-1			
	323	Waste Removal/Disposal	3 -1			
	324	Coal	3-2			
3.3	ONE	MILLION TONNE CASE	3-3			
	331	Introduction	3-3			
	332	Pit Boundaries	3-3			
	333	33 Material Movement				
	335	Mine Equipment	3-3			
	336	Mine Labour	3-4			
3.4	FIVE	3-4				
	341 Introduction					
	342	2 Pit Boundaries				
	343	Material Movement	3-5			
	344	Mine Plan Details	3-5			
		.1 Overview	3-5			
		.2 Top Soil	3-5			
		.3 Waste	3-5			
		.4 Coal	3-6			
		.5 Preparation Plant Coarse Rejects	3-6			
	345	Mine Equipment	3-6			
	346	Mine Labour	3-6			

Table

- 3.1 Material Movement Schedule 1 Mtpy Case
- 3.2 Mine Major Equipment Schedule 1 Mtpy Case
- 3.3 Mine Auxiliary Equipment Schedule 1 Mtpy Case
- 3.4 Mine Labour Requirements 1 Mtpy Case
- 3.6 Material Movement Schedule 5 Mtpy Case
- 3.7 Mine Major Equipment Schedule 5 Mtpy Case
- 3.8 Mine Auxiliary Equipment Schedule 5 Mtpy Case
- 3.9 Mine Labour Requirements 5 Mtpy Case

Figure

- 3.1 Boundary of Areal Extent of Mining Area at Pit Bottom 1 Mtpy Case
- 3.2 1 Mtpy Longitudinal Section of Mine showing yearly progression
- 3.3 Organization Chart 1 Mtpy Case Mining
- 3.4 Boundary of Areal Extent of Mining Area at Pit Bottom 5 Mtpy Case
- 3.5 5 Mtpy Longitudinal Section of Mine showing yearly progression
- 3.6 5 Mtpy Longitudinal Section of Mine showing backfill progression
- 3.7 Organization Chart 5 Mtpy Case Mining
- 3.8 Wing Waste Dump Construction
- 3.9 Mine Progression at Year 5 5 Mtpy Case
- 3.10 Mine Progression at Year 10 5 Mtpy Case
- 3.11 Mine Progression at Year 15 5 Mtpy Case
- 3.12 Mine Progression at Year 20 5 Mtpy Case

Plate

- 3.1 Hobbit-Broatch Resource Area Mine General Arrangement -1 Mtpy Case
- Hobbit-Broatch Resource Area Mine General Arrangement 5 Mtpy Case

PART 4 - MINESITE INFRASTRUCTURE

CONTENTS

()

÷

()

			Page	
4.1	INTR	RODUCTION	4-1	
4.2	ROADS			
	421	Haul Roads	4-1	
	422	Access Roads	4-1	
	423	Ramps	4-1	
4.3	DRA	INAGE	4-1	
	431	Surface	4-1	
	432	Pit Drainage	4-2	
4.4	SER	VICES	4-2	
	441	Water Supply	4-2	
	442	Electrical Power	4-2	
	443	Shops and Warehouse	4-2	
	444	Fuel Supply	4-2	
4.5	ADÅ	AINISTRATION	4-3	
	451	1 Mtpy Case	4-3	
	452	5 Mtpy Case	4-3	
		N 7		

Table

4.1 Administration Labour Requirements - 1 Mtpy Case and 5 Mtpy Case

Figure

- 4.1 Organization Chart 1 Mtpy Case Administration
- 4.2 Administration Chart 5 Mtpy Case Administration

PART 5 - ASSESSMENT PARAMETERS

MINING

CONTENTS

 \bigcirc

()

 \bigcirc

					Page
5.1	INTR	ODU	ICTION	1	5-1
5.2	LAB	OUR			5-1
	521	Intro	oductio	n	5-1
	522	Pers	sonnel		5-1
		.1	Class	sifications	5-1
		.2	Age	of Equipment	5-1
		.3	Shift	Cycles and Systems	5-1
		.4	Utili	zation	5-2
			.41	Availability	5-2
			.42	Utilization	5-2
		.5	Equi	pment Matching	5-2
5.3	EQU	IPME	ENT		5-5
	531	Intr	oductio	ก	5-5
	532	Min	ing Str	ategy	5-5
		.1	Intro	oduction	5-5
		.2	Wast	te	5-5
			.21	Drilling	5-5
			.22	Blasting	5-5
			.23	Loading	5-5
			.24	Trucking	5-5
			.25	Conveying	5-5
		.3	Coal	l Mining	5-8
			.31	Drill/Blast	5-8
			.32	Loading	5-8
			.33	Trucking	5-8

				Page
		.4	Auxiliary Operations	5-8
			.41 Drainage	5-8
			.42 Road Maintenance and Construction	5-8
			.43 Waste Dump Backfill Levelling	5-8
			.44 Conveyor Slewing	5-8
	533	Equ	ipment Availability/Utilization	5-10
	534	Min	ing Equipment Service Life	5-10
	535	Equ	ipment Productivities	5-10
		.1	Assumptions	5-10
		.2	Waste Loading	5-11
		.3	Coal Loading	5-12
		.4	Drills	5-13
		.5	Trucks	5-14
5.4	MIN	ING		5-15
	541	Mat	erial Properties	5-15
		.1	Swell Factor	5-15
		.2	Coal Loss	5-15
		.3	Out of Seam Dilution	5-15
		.4	Moistures	5-15
		.5	Densities	5-15
	542	Der	ivation of Material Quantities -	
		San	nple Calculations - 5 Mtpy Case	5-16
		.1	Run of Mine Feed	5-16
		.2	Coal In Situ Calculation	5-16
		.3	Preparation Plant Reject Calculation	5-16
	5/13	Ц л.	il roads	5-17
	747	1	Design Specifications	5-17
		•1	Maintenance Criteria	5-17
		•4	maintenance Criteria	2-11

 \bigcirc

)

 \bigcirc

	544	Blasting	5-17		
		.1 Requirements	5-17		
		.11 Waste	5-17		
		.12 Coal	5-17		
	545	Production Reserves	5-17		
		.1 Stripped Coal	5-17		
5.5	HYD	ROLOGICAL AND HYDROGEOLOGICAL	5-18		
	551	Introduction	5-18		
	552	Run Off Protection	5-18		
	553	Drainage	5-18		
	554	Water Quality	5-18		
5.6	GEO	TECHNICAL	5-18		
	561	Introduction	5-18		
	562	Pit	5-18		
	563	Waste Dumps	5-18		
5.7	REC	RECLAMATION			
	571	Introduction	5-19		
	572	Top Soil	5-19		
	573	Replacement	5-19		
	574	Revegetation	5-19		

Page

Table

 \bigcirc

()

5.	1	Staff	Classification

- 5.2 Non-Supervisory Classification
- 5.3 Major Equipment Availability Hours Per Year

VOLUME 3

APPENDIX A

CONTENTS

Plate

 \bigcirc

 \bigcirc

A.1	Geological Cross-Section 5500 (East)
A.2	Geological Cross-Section 5000 (East)
A.3	Geological Cross-Section 4500 (East)
A.4	Geological Cross-Section 4000 (East)
A.5	Geological Cross-Section 3500 (East)
A.6	Geological Cross-Section 3000 (East)
A.7	Geological Cross-Section 2500 (East)
A.8	Geological Cross-Section 2000 (East)

VOLUME 4

COAL PREPARATION FACILITIES

-

MOUNT KLAPPAN COAL PROJECT

VOLUME 4

COAL PREPARATION FACILITIES

TABLE OF CONTENTS

PART

- 1 INTRODUCTION
- 2 DESIGN CRITERIA
- 3 PREPARATION PLANT 5 Mtpy CASE
- 4 RAW COAL HANDLING FACILITY 1 Mtpy CASE

MASTER TABLE OF CONTENTS

APPENDICES

PART 1 - INTRODUCTION

CONTENTS

Page

1.1 SYNOPSIS

 \bigcirc

÷ : .

 \bigcirc

()

1-1

CONTENTS

()

()

 \bigcirc

1

Page

2.1	INTRODUCTION	
2.2	PLANT CAPACITY	2-1
2.3	OUT-OF-SEAM DILUTION	2-1
2.4	RUN-OF-MINE SIZE DISTRIBUTION	2-2
2.5	YIELD RANGE	2-2
2.6	MOISTURE CONTENT	2-3
2.7	YIELD AND QUALITY ESTIMATES	2-3

PART 3 - PREPARATION FACILITY - 5 Mtpy CASE

CONTENTS

()

ł

٠

()

Page

ļ

3.1 OVERVIEW

3.2	COA	3-1	
	311	Data Base	3-1
	312	Compositing Procedures	3-2
	313	Preparation Plant Performance	3-2
	314	Raw and Clean Coal Quality	3-3
3.3	СОА	L HANDLING AND PREPARATION PLANT	3-4
	321	Introduction	3-4
	322	Site Location	3-4
	323	Preparation Plant Circuit Alternatives	3-5
	324	Raw Coal Handling	3-6
	325	Preparation Plant	3-7
	326	Thermal Dryer	3-8
	327	Clean Coal Storage and Loadout	3-9
	328	Refuse Disposal	3-9
	329	Support Facilities	3-10
	3210	Equipment List	3-10
	3211	Labour	3-10

Table

3.1	Theoretical Coal Quality, By Size and By Seam
3.2	Theoretical Coal Quality Summary, By Size and By Seam
3.3	Plant Feed Washability Composite, 10 mm x 0.6 mm Size Fraction
3.4	Plant Feed Washability Composite, 0.6 mm x 0.15 mm Size Fraction



- 3.6 Coal Quality Characteristics, Summary By Seam
- 3.7 Coal Quality Characteristics, By Seam
- 3.8 Estimated Clean Coal Yield and Quality, Partial Cleaning
- 3.9 Estimated Clean Coal Yield and Quality, H.M. Cyclone (plus 0.15 mm) - Flotation
- 3.10 Estimated Clean Coal Yield and Quality, H.M. Cyclone - W.O. Cyclone - Flotation
- 3.11 Estimated Clean Coal Yield and Quality, H.M. Vessel - H.M. Cyclone - W.O. Cyclone - Flotation
- 3.12 Estimated Clean Coal Yield and Quality, Jig-Water Only Cyclone - Flotation
- 3.13 Major Equipment List, 5 Mtpy
- 3.14 Labour Schedule

Figure

- 3.1 Estimated Yield and Quality, Jig-Water Only Cyclone - Flotation
- 3.2 Coal Processing Facility, Site Plan
- 3.3 Estimated Yield and Quality Jig Vs. Heavy Media Cleaning
- 3.4 Plant Organization Chart 5 Mtpy Case

Plate

- 3.2 Preparation Facility Flowsheet
- 3.3 Proposed Preparation Plant Flowsheet
- 3.4 Partial Cleaning Flowsheet
- 3.5 H.M. Cyclone W.O. Cyclone Flotation Flowsheet
- 3.6 H.M. Vessel H.M. Cyclone W.O. Cyclone -Flotation Flowsheet
- 3.7 Preparation Plant Section, Sheet 1
- 3.8 Preparation Plant Section, Sheet 2

PART 4 - RAW COAL HANDLING FACILITY - 1 Mtpy CASE

CONTENTS

4.1	COA	L QUALITY	4-1
4.2	RAW	COAL FACILITY	4-1
	421	Introduction	4-1
	422	Raw Coal Handling	4-1
	423	Support Facility	4-2
	424	Equipment List	4-2
	425	Labour	4-2

Page

Table

4.1	Major Equipment List, 1 Mtpy Case
4.2	Labour Schedule

APPENDICES

TABLE OF CONTENTS

APPENDIX

()

- A SIMULATED EQUIPMENT PERFORMANCES
- B COMPOSITE WASHABILITY DATA

VOLUME 5

.

INFRASTRUCTURE
MOUNT KLAPPAN COAL PROJECT

ſ

VOLUME 5

INFRASTRUCTURE

TABLE OF CONTENTS

PART

- 1 INTRODUCTION
- 2 COAL TRANSPORTATION
- 3 ROAD ACCESS
- 4 SERVICES
- 5 SOCIO ECONOMIC IMPACT
- 6 ENVIRONMENT

REFERENCES

MASTER TABLE OF CONTENTS

PART 1 - INTRODUCTION

CONTENTS

Page

1.1 SYNOPSIS

1-1

Plate

 \bigcirc

1.1 Regional Access Map

PART 2 - COAL TRANSPORTATION

CONTENTS

2.1	INTF	RODUCTION	2-1
2.2	RAII	L HAULAGE	2-1
	221	Overview	2-1
	222	Main Line Completion	2-1
	223	New Railway Cutoff- Southern Route 1	2-1
	224	New Railway Cutoff - Alternative Southern Route 1A	2-2
	225	New Railway Cutoff- Northern Route 2	2-2
	226	Railway Design Criteria	2-3
2.3	POR	RΤ.	2-3
2.4	SLU	RRY PIPELINE	2-4

Plate

2.1	Key Map - Rail Access Routes
2.2	Proposed Continuation of B.C. Railway
2.3	Railway Link Between BCR & CNR Routes 1 & 1A
2.4	Railway Link Between BCR & CNR Routes 2 & 2A

PART 3 - ROAD ACCESS

CONTENTS

 \bigcirc

()

3.1	INTI	RODUCTION	3-1
3.2	PER	MANENT ROAD ACCESS	3-1
	32 1	Route 1	3-1
	322	Route 2	3-1
	323	Permanent Road Design Criteria	3-1
3.3	TEM	IPORARY ACCESS ROAD	3-2
	331	Route and Requirements	3-2

Plate

3.1	Key Map - Road Access Routes
3.2	Preliminary Access Road
3.3	New Highway - Route 1
3.4	New Highway - Route 2

PART 4 - SERVICES

Page

CONTENTS

 \bigcirc

()

()

4.1	INTRODUCTION	4-1
4.2	ACCOMMODATIONS - 1 Mtpy CASE	4-1
4.3	ACCOMMODATIONS - 5 Mtpy CASE	4-1
4.4	ELECTRICAL POWER	4-1
4.5	WATER	4-2

PART 5 - SOCIO-ECONOMIC IMPACT

CONTENTS

ŗ

5.1	INTF	RODUCTION	5-1
	511	Purpose	5-1
	512	Impact Area	5-1
	513	Study Approach	5-2
	514	General Assumptions	5-2
5.2	THE	MOUNT KLAPPAN PROJECT	5-3
	521	Location and Access	5-3
	522	Resource Potential, Quality and Marketing	5-4
	523	Production Alternatives	5-4
	524	Timing of Development	5-5
	525	Infrastructure Requirements	5-5
	526	Socio-Economic Impact Potential	5-6
5.3	ТНЕ	NOR THWEST REGION OF BRITISH COLUMBIA	5-8
	531	Overview	5-8
	532	Present Socio-Economic Conditions	5-9
		.1 Population and Workforce	5-9
		.2 Present Industrial Activity	5-10
		.3 Income Levels	5-11
	533	Existing Infrastructure	5-12
		.1 Roads	5-12
		.2 Railway	5-13
		.3 Air	5-14
		.4 Port Facilities	5-14
		.5 Electricity	5-15
		.6 Industry	5-16
	534	General Development Potential	5-16

Page

	535	Deve	lopment Constraints	5-17
		•1	Southern Sector of the Impact Area (South of Hazelton)	5-17
		.2	Northern Sector of the Impact Area (North of Hazelton)	5-18
	536	Prov to D	incial Government Attitude evelopment	5-19
E 1.	500		ONOMIC INDACTS I NEW DOODUCT COM	5 10
5.4	SOC.		ONOMIC IMPACTS - 1 MTpy PRODUCT COAL	5 10
	541	Emp		J-19 5 10
		•1	Direct Regional Employment	2-19
		.2	Indirect Regional Employment	5-20
	542	Regi Salai	onal Personal Income and ry Levels	5-20
	543	Prov	incial Employment and Income	<i>5</i> -20
	544	Regi	onal Population	5-21
	545	Maxi Hous	imum Impacts on Family Structure, sing Requirements, and Community Services	5-22
		.1	Family Structure and Population Demographics	5-22
		.2	Housing Requirements	5-24
		.3	Services	5-24
	546	Cons	struction Employment	5 <u>-2</u> 5
	547	Infra	structure	5-26
		.1	Rail	5-26
		.2	Road Access	5-26
		.3	Port	5-27
	548	Expe	enditures	5-27
		.1	Capital Expenditures	5-27
		.2	Operating Expenditures	5-28
	549	Sum	mary	5-28

 \bigcirc

 \bigcirc

 \bigcirc

Page

5.5	5.5 SOCIO ECONOMIC IMPACTS - 5 Mtpy PRODUCT CO						
	551	Empl	Employment				
		.1 Direct Regional Employment					
		.2	Indirect Regional Employment	5-30			
	552	Regi	onal Personal Income and Salary Levels	5-30			
	553	Provincial Employment and Income 5-30					
	554	Regi	onal Population	5-31			
	555	Fami	ily Structure and Population Demographics	5-31			
	556	Housing Requirements					
	557	Construction Employment					
	558	Services					
		.1	Education	5-34			
		.2	Medical and Health	5-34			
		.3	Government Services	5-34			
		.4	Commercial Facilities	5-34			
	559	Infrastructure					
	5510	Expe	enditures	5-36			
		•1	Capital Expenditures	5-36			
		.2	Operating Expenditures	5-37			
	5511	Sumi	mary	5-37			
			-				

5.6 AREAS FOR FURTHER STUDY 5-38

Table

5.1	Population Model for Mount Klappan Coal Project (1.0 Million Tonnes)

5.2 Population Model for Mount Klappan Coal Project (5.0 Million Tonnes)

Plate

5.1 Impact Area - Mount Klappan Coal Project

PART 6 - ENVIRONMENT

CONTENTS

_ _ _ _

 \bigcirc

()

C

. .

6.1	SUM	MAR	Y		6-1
6.2	DISC		NE-SP	ECIFIC DISCUSSION	6-1
	621	LAN		E	6-1
		.1	Intro	oduction	6-1
		.2	Reso	ource Use	6-1
			.21	Mining	6-1
			.22	Forestry	6-1
			.23	Agriculture	6-2
			.24	Recreation	6-3
		.3	Land	d Status	6-3
			.31	Government/Residential/	6-3
				Commercial	
			.32	Native	6-4
			.33	Historical	6-4
	622	PH	SICA	L ENVIRONMENT	6-5
		.1	Intro	oduction	6-5
		.2	Atm	ospheric Environment	6-5
6.2			.21	Climate	6-5
			.22	Air Quality and Noise	6-6
		.3	Terr	estrial Environment	6-6
			.31	Geology/Vegetation	6-6
			.32	Wildlife	6-7
		.4	Aqu	atics Environment	6-8

6-9 6.3 PROJECT-SPECIFIC DISCUSSIONS 6-9 631 Introduction 6-9 632 Mine Site Existing Railway Plus Alternative #1 6-10 633 Existing Railway Plus Alternative #2 6-10 634 6-10 635 Slurry Pipeline 6-10 636 Access Road

Table

6.1	Comparative Meteorological Information
	for the Mt. Klappan Coal Property (Didene)
	and Other Western Canadian Reporting Stations

6.2 Tabulation of Environmental Issues Impacting the Mt. Klappan Coal Project

Page

APPENDIX A

 \bigcirc

()

GEOLOGICAL CROSS SECTION

APPENDIX A

CONTENTS

Figure

;

A-3 Geological Cross Section 4500 S (East)



•

.

-

•

· · ·					
				`	
	······································		;		•
	HOBBIT CREEK	B.C.R.			SPATSIZI
	COO 2200 8200			· · · · · · · · · · · · · · · · · · ·	
	8		· · · · · · · · · · · · · · · · · · ·		
				· · · · · · · · · · · · · · · · · · ·	
	A B				
	······································				<u> </u>
· · · · · · · · · · · · · · · · · · ·				·····	•••••••••••••••••••••••••••••••••••••••
I					
	LIMITS OF MINING AREA AT PIT BOTTOM - 5 Mtpy Case				
	•	>-			
·					
				- 1	

.



•

· -

APPENDIX B

()

(

(

REGIONAL MAP

APPENDIX B

- -----

CONTENTS

- - -

- -

Regional Map

(

- . .

· · · · · ·

1



Produced jointly by GULF CANADA DRAFTING DEPT. and HARDY ASSOC. (1978) LTD., MAPPING SECTION. Revised to January, 1983.

LEGEND

-(16)---

-

.

· 6750

∕⊗4

8

Highway Road, proposed ... Road, alternate ... Railway ... Proposed Railway cut-off sines andere series andre stati Alternate Railway cut-off . +---------+ Pipeline, possible . Proposed Dam Site Gulf Property Proposed Pit, Mt. Klappan property Proposed Townsite, Mt. Klappan property ... Boundary, Park or Reserve ... Boundary, International ____ Spot Elevation (feet above sea level) Contours (1000 Foot Interval) ... Producing Mine (see separate list) Prospect . City, Town . •



NORTHWEST BRITISH COLUMBIA

Miles 5

0

REFERENCE NOTE

Producing Mines: from The Northwest Region -B.C. Regional Economic Study, 1982.

Prospects: from Kitimat-Stikine Regional District -1:500,000 Regional Resource Map, 1981. Base Map: from Dept. of Energy, Mines and Resources, Surveys and Mapping Branch, current N.T.S. series maps.

SCALE 1: 500,000 50 Kilor res Kilometres 10 5 0 20 30 40 10 30 Miles ------

20



1 DOME MOUNTAIN - Ag, Pb, Zn 2 DUTHIE - Ag, Pb, Zn, Au, Cd, Cu 3 SILVER STANDARD - Ag, Pb, Zn, Au, Cu 4 KITSAULT - Mo 5 SCOTTIE GOLD - Au, Ag 6 GRANDUC - Cu, Ag, Au

PRODUCING MINES

-

