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FEASIBILITY STUDY
KILBORN ENGINEERING - Dec. 1982

# SCHEDULE "A"

# KILBORN ENGINEERING (B.C.) LTD. FEASIBILITY STUDY

KAMAD SILVER CO. LTD.

HOMESTAKE PROPERTY

KAMLOOPS, B. C.

FEASIBILITY STUDY

# SUBMITTED BY:

KILBORN ENGINEERING (B. C.) LTD. 101 - 1199 West Pender Street Vancouver, B.C. V6E 2R1

# KAMAD SILVER CO. LTD.

# HOMESTAKE PROPERTY

# FEASIBILITY STUDY

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1.0 INTRODUCTION

#### 1.0 INTRODUCTION

This Report has been commissioned by Kamad Silver Co. Ltd., Vancouver, British Columbia. It comprises a review of mineral reserves of the Homestake Deposit, a Preliminary Mine Development Plan, a Mining Plan, Surface Mine Facilities, and processing the ore at the Dekalb Mill. Two concentrates will be produced, one will be a bulk sulfide concentrate containing the silver, gold, copper, lead and zinc values and the other will be a barite concentrate for marketing to the oil drilling industry. The Report has been prepared by Kilborn Engineering (B. C.) Ltd., and is based on site visits to the property, information from suppliers and independent contractors, and information supplied by Kamad Silver Co. Ltd.

This Report may be used by Kamad Silver for any legitimate purpose; but may not be excerpted or condensed without the written consent of Kilborn.

#### 1.1 THE PROPERTY

The property is situated in the Kamloops Mining District of British Columbia in TWP 25, R13 W6 at  $51^{\circ}-10^{\circ}$  north latitude  $119^{\circ}-48^{\circ}$  west longitude. Its location is 54 road miles northeast of Kamloops, British Columbia, a &ity of 65,000 population.

The property consists of a contiguous group of 6 Crown Grant claims and 15 claims located under the modified grid system.

Present claim holdings include the following:

# (a) Crown Grant Claims

Name	<u>Lease No.</u>
Homestake	L827
Maple Leaf	L828
Troublesome	L829
Argentum	L830
Silver Star No. 1	L4567 & L4568
Silver Star Fraction	L4566

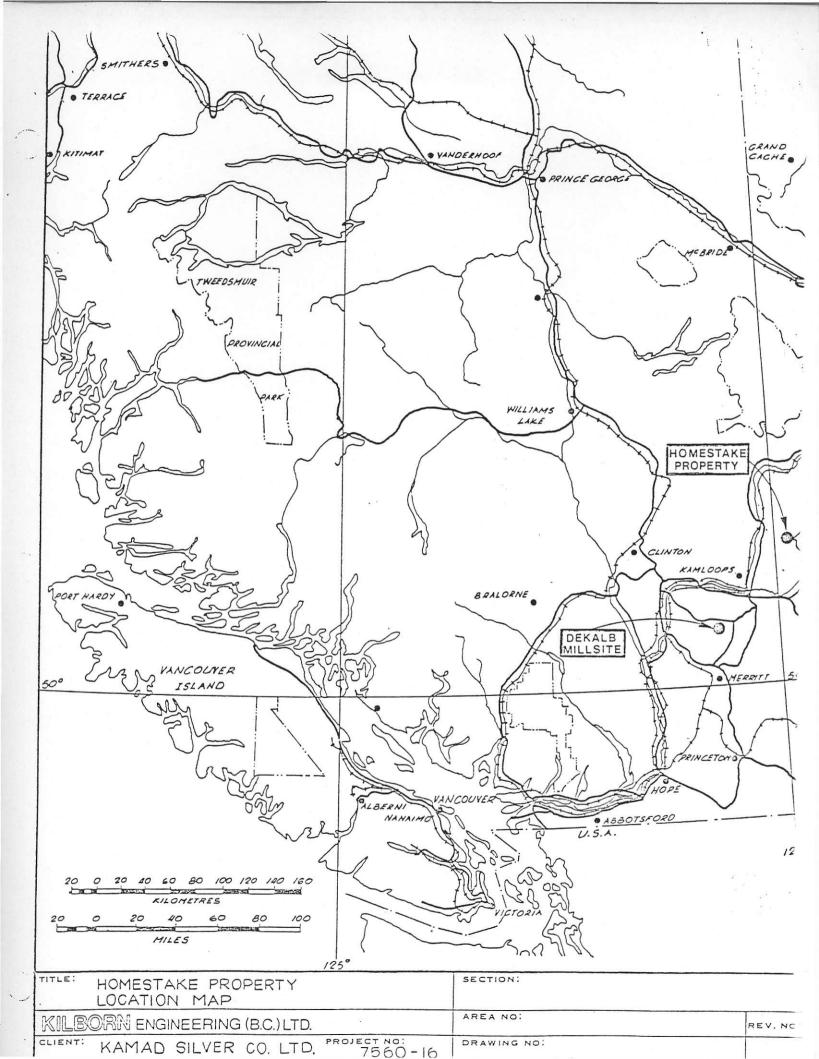
#### (b) Located Claims

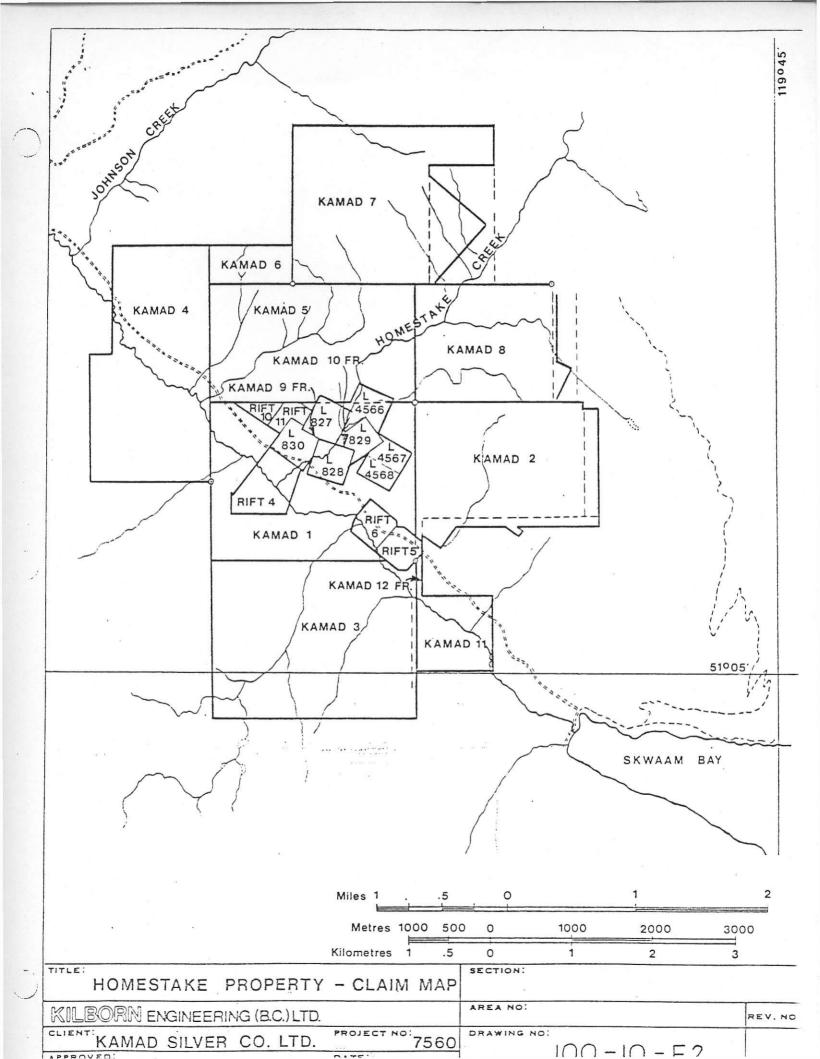
Name			Record No.	Date of	Issu	16
Kamad	1		2685	June	27,	1980
Kamad	2		2686	June	27,	1980
Kamad	3		2687	June	27,	1980
Kamad	4		2688	June	27,	1980
Kamad	5		2689	June	27,	1980
Kamad	6		2690	June	27,	1980
Kamad	7		2691	June	27,	1980
Kamad	8		2692	June	27,	1980
Kamad	9	Fraction	2693	June	27,	1980
Kamad	10	Fraction	2694	June	27,	1980
Kamad	11		3941	February	y 8,	1980
Kamad	12	Fraction	3942	February	y 8,	1980
Rift	4		2383	February	y 5,	1980
Rift	5		2384	February	y 5 <b>,</b>	1980
Rift	6		2385	February	y 5 <b>,</b>	1980
Rift	9		2388	February	y 5 <b>,</b>	1980
Rift	10		2389	February	y 5,	1980

The Claims are registered in the Kamloops Mining Division of British Columbia.

## 1.2 ACCESS

The property is located west of Adams Lake, 19 miles from the community of Barriere. Year-round access is provided by No. 5 Highway from Kamloops to Louis Creek, a distance of 37 miles, thence by gravel road for approximately 17 miles toward Skwaam Bay. From this point, a 0.3 mile sinuous road leads up the north side of Sinmax Creek Valley to the portal at the 1750 foot level; this road continues an additional 1.2 miles to the 2250 foot level portal. Both elevations are distances above sea level.





#### 1.3 TOPOGRAPHY AND CLIMATE

Sinmax Creek flows southeast on the floor of a 'U' shaped valley at approximately 1400 feet above sea level. The valley floor rises gently to 1700 feet above sea level and then more steeply to an elevation of 3000 feet. Peaks in the area rise to approximately 4000 feet. Numerous freshets and creeks are evident on the slopes. Overburden is generally minimal on the slopes, increasing to at least 100 feet in places along the toe of the slopes.

The Sinmax Valley is a semiarid area. Irrigation is practiced by farmers in the valley. Winter temperatures are normally minus  $10^{\circ}$  to minus  $5^{\circ}$  Celsius, however minus  $35^{\circ}$  temperatures have been experienced. Summer highs reach in excess of  $30^{\circ}$  Celsius. The nearest recording station is Barriere, British Columbia (Reference: Environment Canada, Canadian Normals, Precipitation, Temperature, 1941-1971). Recorded precipitation averages 17.5 inches per year.

Water is available from Adams Lake, approximately 4 miles east, or from a well drilled on a nearby farm.

Electrical power in the valley provides rural electrification which is not adequate and would require upgrading if used for a mining operation.

# 1.4 HISTORY

The property has been mined intermittently since the turn of the century. Until 1929, the operation consisted of the recovery of high grade silver ore. In 1935, a mill of 50 tons per day capacity operated for a short period of time.

The existing development consists of approximately 3000 feet of drifting and crosscutting from an adit at the 2250 foot elevation, 1900 feet of crosscutting on the 1750 foot elevation, and approximately 1900 feet of raising above the 2250 foot level and 445 feet of raising from the 1750 foot level to the 2250 foot level.

2.0 <u>SUMMARY</u>

#### 2.0 SUMMARY

The gross receipts for sulphide concentrate were calculated from a recent (August 1981) smelter contract, issued by a smelter in southern British Columbia. Recent discussions with representatives from this smelter have indicated that current specifications may be more stringent than those in the contract. Until negotiations with various smelters have been carried out, the gross receipts shown (for the sulphide concentrate) may be considered only as a guide.

The gross receipts shown for barite concentrate were calculated using a selling price as supplied by Kamad Silver. No detailed investigations were done by Kilborn to verify the selling price, or to verify the capacity of the market to absorb this tonnage of barite, as marketing of barite was not included in the scope of this Study.

#### 2.1 MINERAL RESERVES

The in place mineral reserve of drill indicated quality is:

<u>Vein</u>	Tons	Au oz/.ton	Ag oz/ton	الادب: Pb <u>%</u>	Zn <u>%</u>	Cu %	BaSo <sub>4</sub>
300	100,500	.023	8.53	1.74	2.89	.42	24.9
400	51,300	.008	5.95	.84	1.54	.22	27.5
500	123,900	.015	5.32	1.01	1.89	.20	50.1
TOTAL	275,700	.017	6.61	1.24	2.19	.28	36.7

Mining reserve with dilution of 15 percent and mining recovery of 80 percent is:

Tons	Au oz/ton	Ag oz/ton	Pb %	Zn %	Cu %	BaSO <sub>4</sub>
253,600	.015	5.75	1.08	1.90	.24	31.9
======	====	====	====	====	===	=====

# 2.2 <u>CAPITAL REQUIREMENTS</u>

Capital requirements to bring the mine into production at 8400 tons per month, acquire the Dekalb Mining Corporation assets and remodel the mill are as follows:

Dekalb Purchase	\$ 1,700,000
Mine Surface Facilities	708,400
Mine Development	1,533,800
Mill Modifications	844,200
Engineering	250,000
Contingencies	333,600
TOTAL	\$ 5,370,000

## 2.3 OPERATING COSTS

The operating costs at full production will be:

Administration including ore haulage	\$ 13.43
Mining	30.75
Milling	17.55
Contingency	6.19
TOTAL	\$ 67.92
	=====

# 2.4 GROSS RECEIPTS

The income per ton ore treated is estmated to be as follows:

Silver Bearing Sulphide Concentrate - \$ 59.72 ton ore

Barite Concentrate - \$ 38.38 ton ore

Total Receipts - \$ 98.10 ton ore

TABLE 2-1

CASH FLOW (\$1000)

	[					Year	1							Year	2			Yea	ir 3		Year	. 4
Item	1	2	3	4	5	6	7	8	9	10	11	12	1s t	2nd	3rd	4th	lst	2nd	3rd	4th	lst	2nd
e Pevelopment																						
e Surface Plant																						
l Modification .																h						ļi
e Production	-		. 1400	1100	3500	5100	7800	8100	8400	8400	8400	8400	25200	25200	25200	25200	25200	25200	25200	25200	25200	8400
1 Production	-	-	-	-	1800	8400	8400	8400	8400	8400	8400	8400	25200	25200	25200	25200	25200	25200	25200	25200	25200	8400
:kpile	-	-	1400	2500	4200	900	300	-	-	-	-	-	-	-	-	-	_	-	-	-	-	-
:1pts	-		_		•	177	824	824	824	824	824	824	2472	2472	2472	2472	2472	2472	2472	2472	2472	1628
anditures:	•		!																			
Purchase	1700	-	_	-	-	-	-	-	-	-	-	-	_	-	-	-	-	•	-	-	-	
: Plant	180	180	180	168	-	-	_	-	,	-	-	-	_	-	-	-	-	-	-	-	-	-
Development	250	250	260	260	260	254	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-	-
Modification	94	250	250	250	-	-	-	-	-	-	-	_	-			-			-		-	-
neering	75	75	50	50		-	-	-	-	-		-	-	-	-	-	-	-	-	-	-	-
Ingenotes	60	75	74	74	26	25	-	-	-	-	_	-	-	-	_	-	-	-	-	-	-	-
ations	-	-	-	-	163	264	571	571	571	571	571	571	1712	1712	1712	1712	1712	1712	1712	1712	1712	571
1 Expenditures	-	-	-	-	449	543	571	571	571	571	571	571	1712	1712	1712	1712	1712	1712	1712	1712	1712	571
hly Cash Flow	(2359)	( 830)	(814)	( 802)	( 449)	( 366)	253	253	253	253	253	253	760	760	760	760	760	760	760	760	760	1057
ulative Cash Flow	(2359)	(3189)	(4003)	(4805)	(5254)	(5620)	(5367)	(5114)	(4861)	(4608)	(4355)	(4102)	(3342)	(2582)	(1822)	(1062)	(302)	458	1218	1978	2738	3795

# 2.5 CASH FLOW PROJECTION

The following Table 2-1 gives the cash flow projection for the Project. The payback period is 21 months after start of milling, assuming no interest on capital. Projected cash flow after return of investment, not considering taxes, is \$3,795,000.

## 2.6 COMMENTS

The schedule is contingent on receipt of government permits and approvals without any delays.

3.0 GEOLOGY AND MINERAL RESERVES

#### 3.0 GEOLOGY AND MINERAL RESERVES

#### 3.1 GEOLOGY

The rock units in the area of the Homestake Deposit are members of the Adams Lake Series of the Shuswap Terrain.

Within the immediate mine area, the rocks vary from a moderately competent quartz schist to a highly incompetent quartz-sericite-talc schist. In some areas, stringers and lenses of quartz comprise up to 30 percent of the country rock.

The schistosity strikes northwest and dips from 15 to  $45^{\circ}$  to the northeast.

#### 3.2 MINERALIZATION

There are three tabular mineralized zones which are subparallel to the schistosity. These zones have a strike of north  $40^{\circ}$  west and dip from  $15^{\circ}$  to  $35^{\circ}$  to the northeast. Mineralization within the zones consists of sphalerite, galena, chalcopyrite, tetrahedrite, barite and minor amounts of proustite. The mineralization within the zones occurs in lenses and disseminations.

The three zones, in vertically descending order, are the 300, 400 and 500 zones. The 300 and 400 zones occur as a single zone near the centre of the deposit. Elsewhere, they are separated by up to 40 feet of schist. The 500 zone, which is approximately 120 feet below the 400 zone, consists of two distinct horizons in most of the eastern portion of the deposit. These horizons are separated by 10 to 15 feet of schist. The 500 zone appears as one horizon in the western portion of the deposit.

### 3.3 <u>MINERAL RESERVES</u>

The mineral reserves have been calculated based on information received from Kamad Silver. The method used in the calculation of these reserves was the polygon method. Polygons were measured on a horizontal plane. Drill intersections were converted to vertical thicknesses. The polygons which are included in the reserves, are shown on Drawing Nos. 200-05-F1, 200-05-F2 and 200-05-F3. Around the periphery of the deposit, the polygons are assumed to extend a distance equal to the width of the zone at the nearest known intersection.

The mineral reserves were calculated using a factor of 10 cubic feet per ton. Britton Research carried out Specific Gravity Tests on the metallurgical samples. The Specific Gravity of the sample approximating the grade of the ore reserves is 3.29, which converts to a tonnage factor of 9.74. Using a tonnage factor of 10 gives a small safety factor.

The nature of the mineral occurrences in the Homestake Deposit are such that lenses cannot be developed to the proven category without completely developing the zones for mining.

For the information of the reader, the following definitions from 'Performance Standards for Professional Engineers Advising on and Reporting on Mineral Properties', published by the <u>Association of Professional Engineers of the Province of Ontario</u>, 1969 are:

- (1) 'Ore is a natural aggregate of one or more minerals which, at a specified time and place, may be mined and sold at a profit, or from which some part may be profitably separated.
- (2) Proven Ore or Measured Ore is that material for which tonnage is computed from dimensions revealed in outcrops or trenches or underground workings and/or drill holes and for which the grade is computed from the results of adequate sampling. The sites for inspection, sampling, and measurement are so spaced and the geological character so well defined that the size, shape, and

mineral content are established. The computed tonnage and grade are judged to be accurate within limits which must be stated. It must be stated whether the tonnage and grade of 'proven' or 'measured' ore is in situ or extractable, with dilution factors shown, and reasons for the use of these dilution factors clearly explained.

- (3) Probable Ore or Drill Indicated Ore is that material for which tonnage and grade are computed partly from specific measurements, samples, or production data, and partly from projection for a reasonable distance on geologic evidence. The sites available for inspection, measurement, and sampling are too widely or otherwise inappropriately spaced to outline the material completely or to establish its grade throughout.
- (4) Possible Ore or Inferred Ore is that material for which quantitative estimates are based largely on broad knowledge of the geologic character of the deposit and for which there are few, if any, samples or measurements. The estimates are based on an assumed continuity or repetition for which there are reasonable geological indications; these indications may include comparison with deposits of similar type. Bodies that are completely concealed may be included if there is specific evidence of their presence.'

Considering the above definitions, the in-place undiluted <u>'drill</u> indicated' mineral reserves are as follows:

		Location		
<u>Item</u>	300 Vein	400 Vein	500 Vein	Total
Tons	100,500	51,300	123,900	275,700
Gold oz/t	0.023	0.008	0.015	0.017
Silver oz/t	8.53	5.95	5.32	6.61
Lead %	1.74	0.84	1.01	1.24
Zinc <sup>,</sup> %	2.89	1.54	1.89	2.19
Copper %	0.42	0.22	0.20	0.28
Barite %	24.9	27.5	50.1	36.7

Dilution from mining will be in the order of 15 percent at no grade. Mining recovery is estimated to be 80 percent of the deposit. Therefore, the mining reserve is as follows:

Quantity	-	253,600	Tons	
Gold	-	0.015	Ounces Troy per Ton	
Silver	-	5.75	Ounces Troy per Ton	
Lead	<b>-</b>	1.08	Percent	
Zinc	-	1.90	Percent	
Copper	-	.24	Percent	
Barite	-	31.91	Percent	

Table 3-1 lists the quantity and grade of the blocks considered within the reserve.

TABLE 3-1

DRILL INDICATED RESERVES

(In-Place Undiluted)

Vein	Polygon	Volume	Tons	Au Oz/t	Ag Oz/t	Pb %	Zn %	Cu %	BaSO <sub>4</sub> 5
300	A	80,000	8,000	.016	3.09	.82	1.43	.19	9.5
	В	177,650	17,800	.007	4.58	.92	1.55	.13	41.0
	С	268,800	26,900	.022	11.87	2.13	3.68	.70	46.8
	D	55,250	5,500	.032	1.28	1.31	2.47	.08	TR
	Ε	30,600	3,100	.072	7.00	7.62	10.59	1.14	TR
	F	135,600	13,600	.059	12.87	3.19	6.26	.56	TR
	G	154,200	15,400	.017	12.25	1.06	.94	.44	13.0
	Н	63,840	6,400	.001	4.50	.57	.83	.10	26.1
	I	18,000	1,800	.014	2.39	.34	.64	.06	19.0
	J	20,000	2,000	TR .	2.89	TR	TR	TR	19.8
TOTAL			100,500	.023	8.53	1.74	2.89	.42	24.9
400	Α	49,150	4,900	.020	5.59	.78	1.63	.17	46.7
	В	227,650	22,800	.006	3.52	.84	1.89	.14	31.8
	С	60,200	6,000	.015	8.39	1.15	1.68	.46	TR
	D	6,130	600	TR	4.76	.84	.92	.16	TR
	Ε	90,500	9,000	.003	7.06	.76	.71	.15	32.5
	F	79,900	8,000	.010	10.10	.75	1.37	.36	20.5
TOTAL			51,300	.008	5.95	.84	1.54	.22	27.5
500	А	250,300	25,000	.010	3.21	.93	1.83	.16	33.0
	В	165,920	16,600	.011	5.69	.80	1.51	.19	57.3
	С	499,910	50,000	.010	4.46	.88	1.81	.18	51.5
	D	92,320	9,200	.013	5.88	.76	1.10	.19	48.7
	Ε	201,360	20,100	.038	8.96	1.69	2.69	.28	70.2
	F	29,760	3,000	TR	8.98	1.42	2.87	.28	TR
TOTAL			123,900	.015	5.32	1.01	1.89	.20	50.1
TOTAL	PROBABLE	RESERVES	275,700	.017	6.61	1.24	2.19	.28	36.7

4.0 MINE SURFACE PLANT

#### 4.0 MINE SURFACE PLANT

The mine surface facilities will consist primarily of a mine office/changehouse building, maintenance shop and power/compressor house. The design of these facilities will be based on minimum on-site construction work and maximum portability. The mine surface facilities will be located near the 1750 level portal. Facilities at the upper portal will be limited to a small utility storage building.

## 4.1 <u>MINE OFFICE/CHANGEHOUSE</u> (Drawing No. 400-10-F1)

The office/changehouse will be a transportable factory manufactured structure. The building will be 68 feet in length by 36 feet in width and will contain the following:

- (a) Male changehouse facilities for 35 men;
- (b) Female changehouse facilities for 6 women;
- (c) First Aid Room;
- (d) Mine Rescue Room;
- (e) Lamp Room;
- (f) Offices for the following:
  - Surveyors;
  - Geologist;
  - Clerk;
  - Mine Superintendent;
  - Shift Bosses.

The building will consist of 3 prefabricated 12 foot by 68 foot sections which will be connected on site. The building location is shown on Drawing No. 100-10-F3.

#### 4.2 <u>MAINTENANCE SHOP</u> (Drawing No. 400-10-F2)

The maintenance shop will be 60 feet by 24 feet in plan dimensions with a 12 foot wall. The building will be a double skin insulated metal clad steel frame structure designed and fabricated for ease of erection on site. It will be erected on a concrete slab. A prefabricated 10 foot by 20 foot structure will be supplied for an office and parts storage.

#### 4.3 COMPRESSOR AND POWER HOUSE

The compressor and power house will be a single skin, metal clad steel frame structure of 40 feet by 40 feet in plan dimensions with a 12 foot wall. The building will house the following:

- (a) Three (3) diesel-powered air compressors which currently are located at the Dekalb Mill. The compressors are IR 600 Gyro Flow, IR 900 Gyro Flow, Atlas Copco AR 3L, and have a total capacity of 2100 cfm.
- (b) Two (2) diesel-electric generators, Model D379 with a generating capacity of 350 kilowatts each.

These units will be set on timbers. No concrete foundations will be required.

### 4.4 ORE HANDLING

The ore will be stockpiled and loaded, with a front-end loader, into trucks as required. The loader is currently available from the equipment at the Dekalb Mill site.

#### 4.5 SITE SERVICES

The Site Services required are mine yard lighting, sewage treatment, fire protection and water supply. Watchman Services for the property are provided by a neighbouring farmer who lives in the valley within sight of the property.

Potable water will be provided by a well. Water from the well will be pumped to a head tank located on the hillside, above the mine portal. This tank also will serve for fire fighting reserve.

Sewage treatment facilities will consist of a prefabricated treatment plant which will service the changehouse and office facilities.

## 4.6 <u>2250 PORTAL SERVICES</u>

A 10 foot by 20 foot combined parts storage and lunchroom will be provided for men working from the 2250 level. This will be a transportable, factory built structure.

5.0 THE MINE

#### 5.0 THE MINE

#### 5.1 MINING PLAN

Two adits provide access to the deposits. The upper adit is the 2250 foot level, and the lower adit is the 1750 foot level. The majority of the existing development is on and above the 2250 foot level. There is one raise which has been driven between the two levels and now is partially caved. The 1750 foot level will serve as the main haulage level. The existing raise between the levels will be rehabilitated and divided into two compartments, for use as a ventilation raise (second exit from the mine) and ore pass for material mined on and above the 2250 foot level.

Preproduction development will consist of enlarging the 1750 foot level for truck haulage, rehabilitating and enlarging the 2250 foot level, rehabilitating the 1750 to 2250 raise and development to prepare three stopes for start of mining. Accelerated development will be continued until there are 4 stopes available for mining.

The mining method will be a short-hole, open-stope and rib pillar method with mining advancing up the dip of the deposit. Ore will be removed by slushing downdip to the loading level where the ore will be removed from the bottom of the stope with 2 cubic yard LHD units and hauled to the ore pass.

The zones will be mined in the following sequence. The uppermost ore, the 300 zone, will be mined prior to mining the 400 zone and the 500 zone. If the 300 zone does not overlay the other zones they can be mined at any time.

# 5.2 <u>STOPING METHOD</u> (Drawing No. 200-05-F14)

The ore bodies will be mined by an open-stope and pillar method. Stopes will be 36 feet along strike 160 to 200 feet up the dip. Rib pillars will be 12 feet wide. The stope development sequence will be as follows:

- (a) At the bottom of the stope block a 10 foot high by 12 foot wide ore removal drift will be driven on the hanging wall of the zone.
- (b) A drawpoint will be driven to the footwall of the zone for ore loading.
- (c) A slot raise will be driven from the drawpoint crosscut to the hanging wall contact of the zone.
- (d) A slusher hoist station will be excavated as shown on the drawing and a 25 horsepower three-drum slusher will be installed.
- (e) Near the centreline of the stope, a 5 foot high by 8 feet wide stope raise will be driven along the hanging wall contact of the zone to the top of the stope block.
- (f) The stope raise will join the hanging wall drift for the next group of stopes.

The mining sequence, once the stope development is complete, will depend on the ore thickness. For ore under 12 feet in thickness, mining will be as follows. Starting at the bottom of the stope, the bottom of the raise will be slashed to the footwall of the ore zone. Concurrent with this slashing the ore will be breasted to the pillar lines, using airleg drills and horizontal holes. The stope back will be supported with a 4 foot by 4 foot rock bolt pattern incorporating wire mesh, if required.

For ore over 12 feet in thickness, mining will incorporate two stages. The first stage will be to mine out an 8 foot upper lift similar to the mining of the narrower thickness stopes. The stope back will be supported with rock bolts as previously outlined. The second stage will be to bench out the remaining ore in the floor of the stope. The benching method will depend on the thickness of the remaining ore.

In both mining cases the ore will be scraped downdip to a loading point utilizing 25 horsepower three-drum electric slushers. The ore will be removed from the drawpoint at the bottom of the stope by a 2 cubic yard LHD unit which will transfer it to an ore pass.

#### 5.3 PREPRODUCTION DEVELOPMENT

Preproduction development will consist of the items listed on Table 5-1.

#### 5.3.1 1750 Foot Level

The existing adit will be enlarged to 10 feet in height by 14 feet in width, to permit the use of trucks which currently are part of the Dekalb inventory. Enlarging will be undertaken by slashing both walls as indicated in Sketch No. 5-1. The walls will be supported with rock bolts and sealed with shotcrete to retard deterioration by exposure to the atmosphere. Where warranted, sets will be installed for additional ground support. The crosscut will be extended 280 feet to the 500 zone and a hanging wall drift will be driven on the 500 zone.

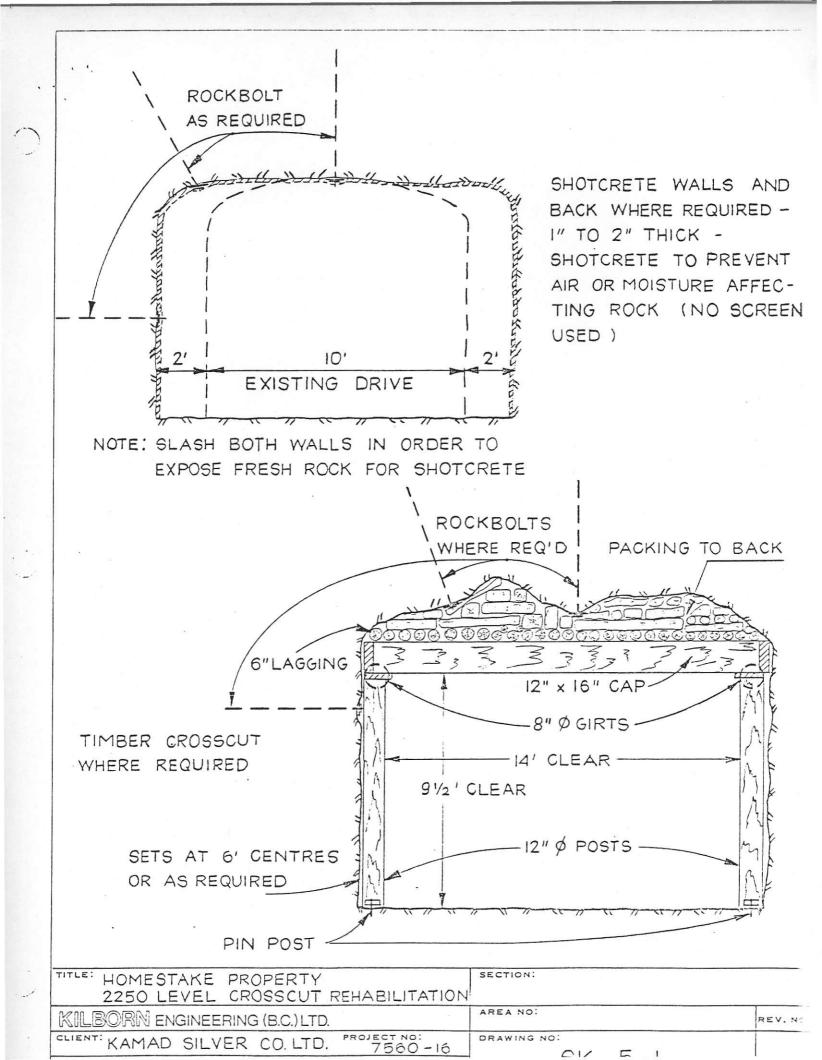
# 5.3.2 2250 Foot Level

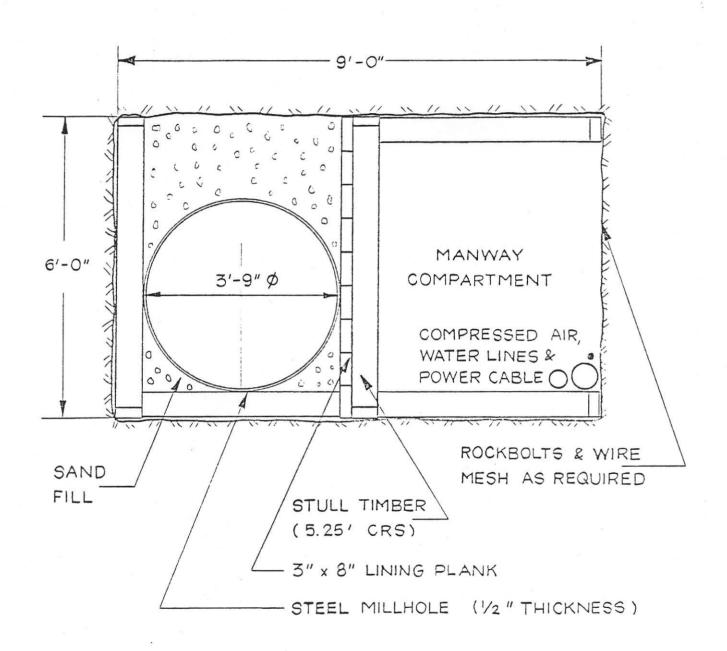
The existing workings will be rehabilitated as required. Headings which are to be used will be enlarged to 9 feet high by 12 feet wide to permit the use of 2-cubic yard scooptrams. The ground will be supported by rock bolts and sets as conditions require. The enlarging will consist of the 2203 drift, 2203 E crosscut and the 2240 drift, as shown on Drawing No. 200-05-F12, for a total of 1040 feet.

From the 2250 foot level, three stopes will be developed.

# 5.3.3 <u>1750 - 2250 Level Raise</u>

The 1750 foot raise, which currently is caved, will be rehabilitated as a combination ventilation raise and ore pass. After the top of the raise has been reached on the 2250 foot level, rehabilitation will be started, working from the top down. The raise will be timbered into two compartments, as shown on Sketch No. 5-2.





CROSS SECTION OF SERVICE RAISE

(NOT TO SCALE)

TITLE: HOMESTAKE PROPERTY	SECTION:	
1750 LEVEL RAISE REHABILITATION		
KILBORN ENGINEERING (B.C.) LTD.	AREA NO:	REV. NO
CLIENT: KAMAD SILVER CO. LTD. PROJECT NO: 7560 -16	DRAWING NO:	

The timber will be installed from the top to the bottom of the raise to support the ground, and the cave material will be removed on the 1750 foot level. After the raise is supported, a steel orepass will be installed in one compartment and a chute will be erected at the bottom of the orepass from which trucks can be loaded.

Stations, for future levels, will be started from the raise as rehabilitation progresses downward so that development may proceed at a future date without damaging the raise system.

### 5.4 <u>VENTILATION</u>

The mine ventilation requirements have been estimated on the basis of the following:

- (a) Working Places (Stopes, Raises, Drifts);
- (b) Total Shift Personnel;
- (c) Diesel Equipment.

The ventilation requirements total 60,000 cubic feet per minute of ventilation air. The main ventilation circuit will be air inflow through the 1750 foot adit, up through the 1750 foot raise and stope workings and out the 2250 foot level. Main ventilation will be provided by a 100 horsepower fan which is part of the Dekalb inventory. Ventilation of the headings and working places which are not on the main ventilation circuit will be by auxiliary fans in bulkheads, with or without ducting as conditions dictate.

Mine air heating will be installed as required. The intent is to use a direct-fired propane mine air heater, located with the main ventilation fan at the portal of the 1750 foot level.

#### 5.5 MINE SERVICES

The major Services to the mine will consist of compressed air, water and electrical power. These Services will be brought into the mine workings through the 1750 foot level adit and up through the 1750 foot raise to the 2250 foot level.

Compressed air capacity will be 2100 cubic feet per minute, supplied by 3 diesel powered compressors. It will be brought to the working areas by a 6 inch diameter pipeline installed in the 1750 foot adit and 1750 foot raise.

Water requirements will be supplied through a 2 inch diameter line paralleling the airline.

Electrical power requirements within the mine are estimated to be 207 kilowatts. This load will consist of:

Ventilation - 100 Horsepower (75 kilowatts);
Pumping - 25 Horsepower (19 kilowatts);
Slushers - 100 Horsepower (75 kilowatts);
Miscellaneous - 50 Horsepower (38 kilowatts).

Power will be brought to the workings and distributed at 600 volts. The material required for power distribution in the mine is in the Dekalb inventory.

## 5.6 ONGOING DEVELOPMENT

Ongoing development during production will consist of the works outlined in Table 5-2.

There will be one two-man raise crew and one two-man drift crew to complete this work. Total development required per day is estimated to be 14 feet.

TABLE 5-2

DEVELOPMENT DURING OPERATIONS

# Stope Development

Type	Size	<u>Quantity</u>
Drifts	9' x 12'	780'
Drawpoints	9' x 9'	420'
Slot Raises	5' x 8'	210'
Manway Raises	5' x 8'	210'
Slusher Stations	8' x 10'	210'
Stope Raises	5' x 8'	3080'
Orepass Raise	5' x 8'	200'
TOTAL LINEAL ADVANCE		5110' ====

## 5.7 <u>EQUIPMENT</u>

The following is a list of mining equipment required for the development and operation of the mine:

Items marked with \* are available from Dekalb inventory.

## 5.7.1 <u>Mining Equipment</u>

<u>Item</u>	Source	<u>Ouantity</u>
Airleg Drill	*	12
Stoper	*	12
Compressors	*	2,100 cfm cap.
Fan	*	100,000 cfm cap.
Fan	*	40,000 cfm cap.
Slusher Hoist	Purchase	4
Scraper	Purchase	4
Sheave Block	Purchase	12
Chain Block	Purchase	4
2 yd <sup>3</sup> LHD	* .	4
Hat Lamp	*	. 40
Utility Hoist	Purchase	3
Pick-Up Truck	*	2
Chain Saw	*	2
Fuel Tank	*	
Freshwater Pump	*	
Wilden Pump	*	4
Mine Rescue Equipment	*	
Pipe Cutter and Stand	*	
Fire Extinguishers	Purchase	
Bit Grinder	*	
First Aid Equipment	Purchase	
Survey Equipment	*	
Hand Tools	*	

#### 5.7.2 Shop Equipment

<u>Item</u>	Source
Jacks	*
Torch	*
Oxygen-Acetylene Carrier	**
Tire Remover	*
Fire Extinguishers	· <b>*</b>
Vices	*
Heaters .	*

## 5.7.3 Office Equipment

The office equipment will come from the Dekalb inventory.

## 5.8 PERSONNEL REQUIREMENTS

<u>Supervisory and Technical Personnel</u> will be kept to a minimum. Requirements are as follows:

- Mine Superintendent;
- Shift Bosses (2 required);
- Geologist;
- Surveyor;
- Survey Helper/Sampler;
- Clerk.

The Mine Production Crew will consist of 31 people. They are distributed as follows:

- (a) Four development miners are required to make up 2 crews. The crews will be required in order to provide 14 feet of daily development advance.
- (b) Sixteen stope miners will be required to operate 4 stopes with 2 man crews working on a 2 shift basis. Approximately 90,000 tons of stope muck are required from stoping; the balance will

be from development. This is equivalent to 360 tons per day, or 22.5 tons per man shift stoping - an acceptable figure.

- (c) Four L.H.D. operators will be required to muck 403 tons per day of ore and a small amount of waste. Although the production of only 101 tons per machine shift is low, the operators will experience higher than average delays on this type of mining and will also look after all nipping in the mine.
- (d) Two truck drivers will be required, one on each shift. Including waste, approximately 425 tons of rock must be hauled daily over 2000 feet.

Cycle time per 12 ton load is calculated to be:

i) Load (mostly from chute, some L.H.D.) - 3 minutes
 ii) Haul 2000 feet at 4 mph average - 6 minutes
 iii) Dump - 1 minute
 iv) Return - 5 minutes
 TOTAL PER TRIP - 16 minutes

Trips per day = 425 ÷ 12 = 35 trips

Time required = 35 x 16 minutes ÷ 60 min/hr = 9.3 hours

On a 2 shift operation, approximately 13.5 hours per day will be available. The 4.2 hours remaining will be occupied by delays, nipping, etc.

(e) In addition, 2 spare miners, one timberman and 2 labourers will be required to look after regular construction, maintenance, ventilation, clean-up, etc.

The Maintenance and Surface Operations crew will consist of 6 people. At least one of the mechanics will work on night shift.

(a) Four mechanical people will be required to look after the diesel mobile equipment as well as to sharpen bits, weld and provide

other normal mechanical services.

(b) In addition, one electrician and one labourer will be employed.

#### 5.9 SUPPLIES AND CONTRACTED SERVICES

Operating supplies required are shown in detail in Section 12.0 - Operating Cost. It may be noted that 50 percent of the total cost of operating supplies is consumed by two items - explosives and diesel oil costs. It may be possible to considerably lower both of these items in the future. If mining conditions permit, ammonium nitrate - fuel oil explosive mixtures may be used in place of water gel explosives. Also, if hydroelectric power becomes available, then diesel oil consumption will drop. However, in order to assure the costs are conservative, neither of the possibilities have been considered.

Certain services, normally performed by mine staff members, will be contracted in order to reduce operating costs. These services include:

- Accounting;
- Secretarial;
- Routine Mining Engineering.

In addition to the above functions, that are normally performed by by the mine staff, other services to be contracted include environmental work and diamond drilling.

Ore handling remains the major contracted service.

6.0 MILL

## 6.0 MILL

## 6.1 <u>INTRODUCTION</u>

Kamad Silver has an option to purchase the assets of Dekalb Mining Corporation. The Dekalb Mill, located in the Highland Valley 51 miles southwest of Kamloops, will be modified to treat 300 dry short tons of ore per day, operating 7 days per week. Allowing for normal operation interruptions, the annual throughput will be 100,800 tons. The mill will be capable of producing two separate concentrates; a bulk sulphide concentrate containing gold, silver, lead, zinc and copper values, and a barite concentrate.

Laboratory testwork results have indicated that the ore can be treated by conventional mineral processing techniques to produce these two concentrates.

For the purpose of this Study, a conventional flowsheet has been developed. The building and most of the major equipment in the Dekalb Mill, including the Crushing Plant, are suitable for the proposed flowsheet.

A market survey of barite based drilling muds indicated that the following characteristics of the barite concentrate produced are required:

- (a) Particle Size:
  - 95 percent or more of minus 325 mesh (44 microns);
  - a maximum of 3 percent plus 200 mesh (74 microns).
- (b) Specific Gravity 4.2 or higher.
- (c) Water Soluble Salts maximum content of 0.1 percent.
- (d) Barimum Sulphate Content minimum 90 percent.
- (e) Moisture Content maximum 2 percent.

### 6.2 REVIEW OF METALLURGICAL TESTWORK

The mill design is based on the test results derived from the reports entitled "Preliminary Flotation Study - Cordilleran Engineering - Homestake Property, Barriere, B.C." dated January 14, 1981 and "Kamad Silver Preliminary Flotation Study II" dated March 13, 1981. Testwork recorded in both reports were carried out by Kamloops Research and Assay Laboratory Ltd. and Met Engineers Ltd.

The plant will be designed to produce bulk sulphide and barite concentrate. The test results indicate that an estimated 300 ppm of mercury can be expected in the bulk sulphide concentrate; possibly making it difficult to market.

Previous testwork carried out by Britton Research Limited in 1973 shows that the barite concentrate with 97 percent barium sulphate, at 34.4 percent plus 325 mesh contains 0.33 percent water soluble salts. If the water soluble salts are interlocked with the barite, regrinding the sulphide tailings prior to barite flotation may upgrade the barite to saleable grade. However, water soluble salts were not in the recent testwork although the ore was ground much finer, consequently it is not certain if the barite concentrate was of marketable grade.

The flowsheet of the process is based on data in the above mentioned report. However, the tests have not been repeated to confirm the results. In order to confirm the test results, and develop revised processes to reduce the mercury content in the bulk sulphide concentrate and the soluble salt content in the barite concentrate, it is recommended that more test work be carried out to ensure that a marketable barite concentrate can be produced.

The following Table 6-1 shows the projected metallurgy of the bulk sulphide and barite concentrates, based on the metallurgical test results and the average grade of the total probable reserves:

TABLE 6-1

PROJECTED MILL FEED AND PRODUCTS

Material	Annual Tonnage	Au (Oz/Ton)	Ag (Oz/Ton)	Pb (%)	Zu <u>(%)</u>	Cu (%)	BaS0 <sub>4</sub> (%)4
0re	100,800	0.015	5.75	1.08	1.90	0.24	31.9
Sulphide Concentrate	6,310	0.14	82.7	11.9	27.3	2.7	5.0
Barite Concentrate	28,428	-	-	-	-	-	93.0

# Metal Recovery

Sulphide Concentrate

Gold	58	percent
Silver	90	percent
Lead	69	percent
Zinc	90	percent
Copper	70	percent

Barite Concentrate

Barium Sulphate 82 percent

## 6.3 PROCESS DESIGN CRITERIA 6.3.1 Crushing Plant Capacity 100,800 tons per year 50 tons per hour Operating Hours Per Day (5 Days per Week) minus 3/8 inch Product Size 6.3.2 Concentrator Capacity 100,800 tons per year 12.5 tons per hour 95 percent Designed Availability Operating Days Per Year 355 Operating Hours Per Day 24 Ore Grades 0.24 percent copper 1.08 percent lead 1.90 percent zinc 31.9 percent barite 5.75 ounces per ton silver 0.015 ounces per ton gold Sulphide Concentrate 6.3.3 2.6 percent copper 10.9 percent lead 27.5 percent zinc 3.5 percent BaSO, ounces per ton silver 0.13 ounces per ton gold 6.3.4 Barite Concentrate 93 percent barite Annual Production of Sulphide Concentrate 6,310 tons Annual Production of Barite Concentrate 28,428 tons

#### 6.3.5 Grinding

## Primary Grinding

Theoretical

Installed

12.5 tons per hour Feed Rate Feed Size minus 3/8 inch 76 percent - 200 mesh Discharge Calculated Horsepower 281 Installed Horsepower 300 7.5 feet diameter x 10 feet Ball Mill Size 10 kilowatt hours per ton Work Index 6.3.6 Barite Regrinding Regrind Ball Mill Size 4 feet diameter x 6 feet Calculated Horsepower 88 Installed Horsepower 100 6.3.7 Flotation Sulphide Conditioning Time: 5.0 minutes Theoretical 15.0 minutes Installed Sulphide Rougher Time: Theoretical 16.8 minutes 22.3 minutes Installed Sulphide Scavenger Time: Theoretical 4.8 minutes 9.5 minutes Installed Sulphide First Cleaning Time: 19.2 minutes Theoretical 21.0 minutes Installed Sulphide Second Cleaning Time: Theoretical 14.4 minutes Installed 15.8 minutes Sulphide Third Cleaning Time:

9.6 minutes.

15.8 minutes

Sulphide Fourth Cleaning Time:					
Theoretical	9.6	minutes			
Installed	16.2	minutes			
Barite Conditioning Time:					
Theoretical	7.2	minutes			
Installed	13.6	minutes			
Barite Rougher Time:					
Theoretical	14.4	minutes			
Installed	19.8	minutes			
Barite Scavenger Time:					
Theoretical	4.8	minutes			
Installed	9.8	minutes			
Barite First Cleaning Time:					
Theoretical	12.0	minutes			
Installed	12.8	minutes			
Barite Second Cleaning Time:					
Theoretical	7.2	minutes			
Installed	7.8	minutes			
Barite Third Cleaning Time:					
Theoretical	7.2	minutes			
Installed	7.3	minutes			
Thickening and Filtration					
Sulphide Concentrate Thickening Area:					
Theoretical	6.0	ft <sup>2</sup> per	ton	per	day
Installed		ft <sup>2</sup> per			
Sulphide Concentrate Filter Unit Area:				•	•
Theoretical	4	ft <sup>2</sup> per	ton	per	day
Installed		ft <sup>2</sup> per			
Barite Concentrate Thickening Unit Area:					
Theoretical	6.0	ft <sup>2</sup> per	ton	per	day
Installed		ft <sup>2</sup> per			
Barite Concentrate Filter Unit Area:		•		•	•
Theoretical	5	ft <sup>2</sup> per	ton	per	day
Installed		ft <sup>2</sup> per		•	•
		•		•	

6.3.8

#### 6.4 PROCESS DESCRIPTION

### 6.4.1 Crushing

Run-of-mine ore will be stored in the 150 ton coarse ore bin. The ore will be withdrawn from the bin by a 36 inch wide conveyor belt discharging to a grizzly with 3 inch openings. The oversize will be fed to a 20 inch by 36 inch jaw crusher, set at  $1\frac{1}{2}$  inches. The product, combined with the grizzly undersize, will be conveyed to a 4 foot by 10 foot vibrating double deck screen, with 3/8 inch openings in the bottom deck, in closed circuit with a 4 foot short-head cone crusher. The screen undersize will be conveyed to a 1000 ton fine ore bin. An electro-magnet will be installed ahead of the screen to remove most of the tramp metal.

### 6.4.2 Grinding

Fine ore will be fed from the bin by means of an 18 inch wide belt conveyor to a  $7\frac{1}{2}$  foot diameter by 10 foot long ball mill operating in closed circuit with a 15 inch cyclone. The cyclone overflow, regulated at 76 percent minus 200 mesh, will gravitate to the flotation area. Duplicate cyclones and cyclone feed pumps will be provided. Soda ash will be added in the mill to control the pH of the pulp.

## 6.4.3 <u>Sulphide Flotation</u>

The cyclone overflow will be fed to a sulphide conditioner where the pulp will be mixed with reagents including collectors, frother, modifiers and gangue depressant. From the conditioner, the pulp will flow to a sulphide rougher-scavenger tank, comprising 10 units of Denver No. 21 Sub A flotation cells. The rougher concentrate will be upgraded in four stages of cleaning in a bank of 9 units of Denver No. 18 Sub A flotation cells. The concentrates and the tailings in the cleaning stages will be counter-current and with the final concentrate pumped to the sulphide concentrate thickener. The tailings from the first stage of cleaning combined with the rougher tailings will be fed to the scavenger cells. The scavenger concentrate will be returned

to the grinding circuit, while the tailings will gravitate to the barite regrind circuit.

### 6.4.4 Barite Flotation

The sulphide scavenger tailings will be reground to 80 percent minus 270 mesh, in a 4 foot diameter by 6 foot long ball mill in closed circuit with a 10 inch cyclone. The cyclone overflow will be mixed with barite flotation reagent in a 7 foot diameter by 7 foot high conditioner. After conditioning, the pulp will be fed to a barite rougher-scavenger tank, consisting of 10 units of Denver No. 21 Sub A flotation cells. The rougher concentrate will be upgraded in three stages of cleaning comprising 11 units of Denver No. 18 Sub A flotation cells, with the final concentrate fed to the barite thickeners. The tailings from the first cleaning stage will be combined with the rougher tailings to feed the scavenger cells. The scavenger concentrate will be returned to the barite regrind circuit, while the tailings combined with the overflow from the thickener will be pumped to the tailings impoundment area.

## 6.4.5 <u>Dewatering</u>

The sulphide concentrate will be fed to a 14 foot diameter by 8 foot thickener. The thickener underflow will be filtered by a 6 foot diameter disc filter. The filter cake, with a 12 percent moisture content, will be stored on the sulphide concentrate pad. The filtrate will be returned to the thickener.

The barite concentrate will gravitate to a 28 foot diameter by 12 foot thickener. The thickener underflow will be filtered by a 6 foot diameter disc filter. The filter cake, with 12 percent moisture, will be stored on the barite concentrate pad. The filtrate will be returned to the barite thickener.

The overflow from both thickeners will be combined with the mill tailings and pumped to the tailings impoundment area.

## 6.5 TAILINGS DISPOSAL

The tailings produced by the processing plant will be pumped to the existing tailings impoundment area.

The decant water from tailings storage area will be returned to the mill for re-use in the processing plant.

7.0 ADMINISTRATION

#### 7.0 ADMINISTRATION

#### 7.1 ORGANIZATION AND PERSONNEL

The Mining Department and the Milling Department will report separately to a Mine Manager. The Mine Manager will reside in Kamloops, British Columbia, so he will be available to go to either the mine or the mill, as required. A telephone answering service in Kamloops will be advised daily of the Mine Manager's schedule, and will record messages for him. It is probable that the Mine Manager will perform his routine administrative work in the existing Mine Manager's office in the Dekalb Mill.

Central office functions will be kept to a minimum by:

- (a) Contracting of Services Accounting, Engineering, Secretarial.
- (b) Contracting of Work Ore Haulage, Road Maintenance.
- (c) Assigning the Following Functions as Departmental Responsibilities:
  - i) Safety Supervision and Records;
  - ii) First Aid and Mine Rescue;
  - iii) Employee Relations and Personnel;
  - iv) Purchasing and Warehousing;
  - v) Maintenance;
  - vi) Security.

Table 7-1 is the Organization Chart for the Administration. It is indicated that the Mine Manager will take direct charge of contracted services and contracted work.

The Mine Manager will be the only employee classed as 'Administration'.

## 7.2 SERVICES AND SUPPLIES

## 7.2.1 Secretarial

A secretarial service in Kamloops will be retained to process the Mine Manager's correspondence and files.

## 7.2.2 Accounting

An accounting firm in Kamloops will be retained to handle all the accounting work for the operations. This work will include:

- (a) Payroll and Issue of Cheques;
- (b) Accounts Receivable and Deposit of Receipts;
- (c) Accounts Payable and Issue of Cheques;
- (d) Financial Statements;

Corporate accounting may be integrated with the foregoing, if convenient. Auditing will be a separate function.

### 7.2.3 Engineering

Mining Engineering will be carried out, as required, using the services of a semiretired consulting engineer residing in the Kamloops area or the services of an established engineering firm. Assuming that the position of Mine Superintendent is held by a qualified mining engineer, outside Mining Engineering Services are estimated to be required for one week per month.

Metallurgical Consulting Services will be used on an ongoing basis, probably for 2 days per month.

Environmental Engineering Services will be retained to carry out the monitoring of waste discharges, at the intervals as specified by legislation.

### 7.2.4 Contracted Work

A Contract will be let for trucking the ore from the mine to the Dekalb Mill, a distance of 105 miles.

The ore will be dumped into the receiving bin at the Dekalb Mill. This bin arrangement is capable of receiving 'rear dump' trucks only, and there will be no attendant on duty at the dump.

Truck loads should be weighed at a supervised location, if possible, for payment purposes. However, the weight recorded in the mill could be used as a monthly check, if a satisfactory truck weighing system cannot be set up. Before the Contract is let, a procedure, that can be audited, will be established for recording the tonnage of ore hauled.

As part of the ore haulage Contract, delivery of samples from the mine to the mill will be included.

8.0 INCOME

#### 8.0 INCOME

#### 8.1 INCOME SUMMARY

Income from the Homestake property will be from the sale of concentrates produced by treatment of ore in the modified Dekalb Mill.

Production per month will be:

Sulfide Concentrate - 525.83 tons
Barite Concentrate - 2369.00 tons

from treating 8400 tons of ore.

Income from concentrates, based on \$10.00 U.S. per ounce for silver and barite concentrate of \$136.08 per ton, would be as follows:

Sulfide Concentrate \$ 501,612 per month - \$ 59.72 per ton ore

Barite Concentrate \$ 322,368 per month - 38.38 per ton ore

TOTAL \$ 823,980 per month - \$ 98.10 per ton ore

It is assumed this income will be received approximately 30 days after production of the concentrates.

## 8.2 SULFIDE CONCENTRATE NET SMELTER RETURN

## Sulphide Concentrate

It is assumed that the sulphide concentrate will be sold as lead concentrate, based on Cominco open contract.

#### Metal Prices

Canadian \$ 0.26518 per pound of lead (Northern Miner, Average Canadian \$ 0.3875 per pound of zinc for Nov. 23 to Nov. 29, Canadian \$ 0.69 per pound of copper 1982, 5 days of Quotes)

Canadian \$ 9.8784 per ounce of silver Canadian \$ 12.348 per ounce of silver (Selected) Canadian \$543.312 per ounce of gold  Lead Payment Per Ton of Concentrate		
(11.9 - 0.1 (2.7 - 0.75)) 0.92 x C\$ (0.26518 - 0.1) x 20	C\$	33.024
Zinc Payment Per Ton of Concentrate	-4	000021
(27.3 x 20 - 20) x 0.6 x C\$ (0.3875 - 0.15)	C\$	74.955
Silver Payment Per Ton of Concentrate		
At C\$ 9.8784 per ounce (US \$ 8 per ounce) (82.7 - 0.2 x 2.7) 0.93 x C\$ 9.8784 x 0.97	C\$	732.1528
At C\$ 12.348 per ounce (US \$ 10 per ounce)	C\$	915.1910
Gold Payment Per Ton of Concentrate		
(0.14 - 0.03) x 0.98 x 543.312	C\$	58.5690
Total Payment Per Ton of Concentrate		
Silver at C\$ 9.8784 per ounce (US \$8.00)	C\$	911.8360
C\$ 12.3480 per ounce (US \$10.00)	C\$	1094.8742
Deduction Per Ton of Concentrate		
Treatment Charge	C\$	85.0000
Arsenic and Antimony (1.62 + 0.18 - 0.5) x 1.75)	C\$	2.2750
Alumina 0.9 (2.34 - 0.5)	C\$	1.6560
Moisture 0.4 x 2 + 1 x 2 (12 percent moisture)	C\$	2.8000
Labour Rate (1707 - 1567) x 0.08	C\$	11.2000
Truck Receipt Charge	C\$	6.0000
TOTAL DEDUCTION	C\$ _	108.931

## Concentrate Trucking to Smelter

Per Ton of Concentrate C\$ 32.00

### Net Smelter Return Per Ton of Sulphide Concentrate

At C\$ 9.8784 per ounce of silver (US \$8.00) C\$ 770.905 At C\$ 12.348 per ounce of silver (US \$10.00) C\$ 953.9432

## 8.3 BARITE SALES

Barite will be sold through a drill-mud supply company who will receive the material at the mill in bulk form. Net receipts to Kamad are estimated to be \$136.08 per ton of concentrate.

9.0 SOCIOECONOMIC CONSIDERATIONS

#### 9.0 SOCIOECONOMIC CONSIDERATIONS

## 9.1 SOCIAL IMPACT

It is anticipated that the workforce of 44 persons required at the mine site will be available from the nearby established communities of Louis Creek, Barriere and possibly Kamloops, British Columbia. The workforce of 31 persons required at the mill site will be available from the established communities of Logan Lake and Ashcroft.

The Kamad Silver Project will provide employment for a small but regular workforce. Employment at the mining and milling operations will enable many employees to learn new, specialized skills.

### 9.2 LAND USE

### 9.2.1 Mine Site Area

The existing cleared and levelled area adjacent to the 1750 foot mine portal will be utilized to locate the required surface plant and office structures. The waste rock produced by the mine development work will be utilized to enlarge the plant site area and for surfacing material. Drawing No. 100-10-F3 shows the surface plan of the required mine site buildings and installations.

### 9.2.2 Mill Site Area

No additional site clearing or surface installations are required at the Dekalb mill-site area.

## 9.3 MINE WASTE ROCK MANAGEMENT AND MILL TAILINGS DISPOSAL

## 9.3.1 Mine Waste Rock

The waste rock produced by the 1750 foot level mine development work will be utilized to enlarge the plant site area adjacent to the 1750

foot level portal. Waste rock from the 2250 foot level development work will be placed adjacent to the upper mine portal to provide a suitable service area.

### 9.3.2 Mill Tailings Disposal

The existing Dekalb mill tailings storage area is adequate to accommodate the tailings produced by the processing of the known Homestake mine ore.

## 9.4 WASTE WATER MANAGEMENT

The limited mine discharge water from the 1750 and 2250 foot levels will be collected and discharged to the surface drainage system. If required, the mine drainage will be treated before discharge.

10.0 SCHEDULES

## 10.0 SCHEDULES

The mine development schedule is included as Drawing No. 200-26-F1.

Once government permits have been obtained, the mine development is the critical item in the implementation at the project.

The critical item at the start of the project is the enlarging and support of the 1750 level adit. All development and stoping has been scheduled on a 3 shift per day basis until there are 4 developed stopes, at which time mine operations will revert to 2 shifts per day, 5 days per week.

11.0 CAPITAL COST

#### 11.0 CAPITAL COST

### 11.1 SUMMARY

Capital costs are listed below by major items but do not include money spent prior to the date of this Report or money required to obtain approvals and financing:

Description	<u>Capital Cost</u>
Dekalb Purchase	\$ 1,700,000
Mine Surface Facilities	708,400
Mine Development	1,533,800
Mill Modification	844,200
Engineering	250,000
Contingency	333,600
TOTAL	\$ 5,370,000 =======

## 11.2 BASIS OF ESTIMATE

#### 11.2.1 General

The capital cost estimate is based upon fourth quarter 1982 material, equipment and installation rates.

Construction costs at the mine site are based on supply and erection by manufacturer for the prefabricated buildings and other work being undertaken by local contracting firms.

Underground development costs are based on the work being undertaken by Kamad employee and supervised by Kamad staff which will be hired.

Mill modifications costs are based on undertaking the work with the personnel who will operate the mill.

Local contractors will be engaged to supplement the work force as required.

#### 11.2.2 Design and Construction

The capital costs are based upon the mine drawings, process flowsheets and plant complex drawings included in this Feasibility Report; and equipment lists, general arrangment drawings and flowsheets supplied by Dekalb Mining Corporation.

## 11.2.3 Taxes, Duties and Freight

Taxes have been included on materials incorporated into the permanent structural/civil or building portions of the estimate. Federal sales tax on process equipment and material is not included in the estimate, as current regulations allow for tax exemption under qualification and certification of the end use of the equipment.

All federal government import duties, tariffs and levies have been included where applicable.

The cost of freight, transportation and delivery charges of permanent material and process equipment are included with the applicable specific item of capital cost.

## 11.2.4 Engineering and Site Services

Included in this estimate is the cost of engineering, procurement and site services for surface construction. It has been assumed that piping modification and electric modifications in the mill will be field run and that minimal design drawings will be supplied.

## 11.2.5 <u>Contingency</u>

This includes a specific provision for unforseeable elements of cost within the defined Scope of Work for this Project. This value is based on previous experience, relating estimates and actual costs.

## 11.3 ITEMS EXCLUDED FROM THE ESTIMATE

The following items have not been included in the estimate:

- (a) Previous Project Costs;
- (b) Royalties;
- (c) Owner's Administration Expense;
- (d) Working Capital;
- (e) Feasibility Reports;
- (f) Environmental Studies and Submissions;
- (g) Spare parts assumed to be from Dekalb Inventory;
- (h) Interest on borrowed capital.

## 11.4 ESTIMATE DETAILS

The estimate details are given on the following pages:

KILBORN PROJECT ESTIMATE HOMESTAKE CHENT, KAMAD SILVER CO. LTD PROJECT SUMMARY ESTIMATE NO \_ DEC 1987 LABOUR MATERIAL ACCOUNT QUANTITY TOTAL COST DESCRIPTION UNIT TOTAL UNIT UNIT CODE COST COST COST PROJECT SUMMARY 1/8/200 SITE AND GENERAL 100 206200 18300 CHANGEHOUSE OFFICE 118300 220 POWER / COMPRESSOR HOUSE \_230 63400 MAINTENANCE SHOP" 240 76 800 MINE INDIRECT ACCOUNT. 8/0 33 Ba 209 900 MINE SURFACE FACILITIES MINE DEVELOPMENT 250 618 540 CONCENTRATOR MODIFICATIONS 200 30925c 494750 CONCENTRATOR INDIRECT ACCOUNT. 810 CONCENTRATOR 494 750 ENGINEERING AND CONSTRUCTION SERVICES

750

Contingency

KILBORN PROJECT ESTIMATE HOMESTAKE PROJECT\_\_\_\_ LIENT: KAMAD SILVER CO. LTD SHEET No \_ 1 OF 1 SITE HND GENERAL EXTENSIONS \_\_\_\_\_ Dec 1982 LABOUR MATERIAL ACCOUNT DESCRIPTION TYPE QUANTITY TOTAL COST UNIT TOTAL RATE CODE COST COST COST SUMMARY SITE PREPARATION KONDS AND DRAINAGE 154300 WATER SUPPLY 30 100 46 500 120 30500 SEWAGE DISPOSAL 125 3.0 9 0.0 24400 FUEL STORAGE 130 FIRE Projection 140 Communication 150 SURFACE MOBILE EQUIPMENT 26000

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ILBORN PROJECT ESTIMATE PROJECT\_HOMESTAKE PROJ. No. 7560 NT. KAMAD SILVER CO. LTD. FEASIBILITY STUDY SITE AND GENERAL DEC 1982 LABOUR MATERIAL COUNT TYPE DESCRIPTION QUANTITY TOTAL COST TOTAL RATE COST COST COST SITE PREPARATION - AN ALLOWANCE FOR LEVELING Sum AND BULK EXCAVATION TOTAL SITE PREPARATION POARS AND DRAINAGE UPGRADING OF ROAD FROM HIWAY 1600 TO 1750 PORTAL - GRADING AND GRAVEL TOPPING UPGRADING OF ROAD FROM 1750 6500 FT. FORTAL TO 2250 LEVEL FORTAL - GRADING AND GRAVELLING. ( 4 WHEEL DRIVE QUALITY) DRAINAGE DITCH FROM 1750 PORM Fr. 1500 SETTLING GUD - MINE PORTAL  $\gamma p^3$ 220

TOTAL ROADS AND LRAINAGE

KILBORN PROJECT ESTIMATE LIENTI KAMAD SILVER CO. LA AREA No. 100 DEC 1982 LABOUR MATERIAL ACCOUNT TOTAL COST DESCRIPTION . TYPE QUANTITY UNIT UNIT TOTAL HATE COST COST 120 WATER SUPPLY WATER WELL INCLUDING PURCHASE AND ELECTRIC POWER SERVICE Sum Sum PHELINE -WATER TANK AND FINE WATER STORAGE No PIPING VALVES AND FITTINGS PICLUPILG CLORINATION Sum TOTAL - PLANT WATER SUPPLY 125 SEWAGE DISPOSAL SEWAGE TREATMENT PLANT - PECFABRICATED EXCAUATION AND EACKFILL - FLANT AND PIPELINES PIPING VALVES AND FITTINGS

Sum

Eum

DISPOSAL FIELD

TETAL - SEWAGE DISPOSAL

KILBORN PROJECT ESTIMATE JENTI KAMAD SILVER 6.1ts SHEET No. 3 OF 5 DIV. No. \_\_\_\_\_\_\_\_\_\_\_\_\_\_ESTIMATOR\_\_ Site HND GENERAL AREA No 100 DATE DEC 1982 MATERIAL LABOUR CCOUNT QUANTITY TOTAL COST DESCRIPTION UNIT TOTAL RATE CODE FUEL STORAGE DIESEL FUEL TANK TANK EXISTING PUMP C/W MOTOR. PIPING AND VALUES - DISTRIBUTION INSULATION AND HEAT TRACING Sum. ELECTRICAL POWER SERVICE Sum INSTRUMENTATION PROPANE STORAGE TANK. VAFORIZER SYSTEM. No. EXCAUATION, BACKFILL AND GRAVEL FUNDATION FOR STORAGE TOWKS Sum CONTAINMENT DYKE AROUND DIESEL STORAGE TANK Sum PIPING VALVES AND FIFTINGS Propane Distriction SUM

KILBORN PROJECT ESTIMATE CLIENTI KAMAD SILVER CO. LTD SITE AND GENERAL AREA No 100 EXTENSIONS \_\_\_\_\_ ESTIMATE NO \_\_\_\_ DATE \_\_\_\_\_\_ DEC 1982 LABOUR TYPE QUANTITY DESCRIPTION TOTAL COST CODE UNIT TOTAL RATE COST COST 140 FIRE PROTECTION PORTABLE FIRE PROTECTION EQUIPMENT TOTAL - FIRE PROTECTION COMMUNICATIONS Allow FOR EXTENDING TELEPHONE From MAIN HISHWAY total - Communications

KILBORN PROJECT ESTIMATE LIENTI KAMAD SILVER G. Lts. AREA No 100 EXTENSIONS ESTIMATE NO \_\_\_\_ DATE DEC 1982 MATERIAL ACCOUNT LABOUR TYPE QUANTITY UNIT DESCRIPTION TOTAL COST UNIT TOTAL HATE TINUT CODE COST COST COST SURFACE MOBILE EQUIPMENT PICKUP TRUCK. (FROM DEKALB) AMBULANCE ( PICKUP CONVERSION) No. VAN. - TO HAUL MEN TO 2250L No LOADER (FROM PEKALE) No -total - SURFACE MOBILE EQUIPMENT

KILBORN PROJECT ESTIMATE HOMESTAKE PROJECT\_ PROJ. No 7560 WENT: KAMAD SILVER CO. Lts AREA No 220 EXTENSIONS OFFICE / CHANGEHOUSE CCOUNT LABOUR MATERIAL DESCRIPTION QUANTITY UNIT TOTAL COST CODE TOTAL RATE COST COST CHANGEHOUSE OFFICE FACTORY BUILT MODULAR OFFICE CHANGEHOUSE C/W 36'x 68' - WET & DRY LOCKERS - LPG. HEAT - WACHROOM FACILITIES
PURCHASE TRANSFORTAND INSTALLATION EA SITE PREPARATION UTILITY SERVICES Sum OVERHEAD POWER LPG WATER SEWER ETE. OFFICE FURNITURE AND EQUIPMENT 13y 0

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CILBORN PROJECT ESTIMATE PROJECT\_\_\_\_ HOMESTAKE ENTI KAMAD SILVER LO LTD AREA No 230 ESTIMATE NO \_ POWER / COMPRESSOR HOUSE DEC 1982 LABOUR MATERIAL COUNT TYPE QUANTITY TOTAL COST DESCRIPTION UNIT TOTAL RATE UNIT 300: COST COST COST PREFABRICATED STEEL FRAME BUILDING EA 40 LUNG X 40 WIDE AURCHASE DELIVERY AND FRECTION EXCAVATION AND PACKELL Sum REINFORCED CONCRETE ROOR Sum LIGHTING Sum HEATING Sum CCMPRESSORS - Relocate 3 +RUCKING Sum PIPING VALUES AND FIFTINGS - RELOCATE Sum. MND NOW ELECTRIC GENERATORS - Relocate EA - trucking Sum New WIRING Sum Piping valves AND Fiftings - FUST 5,~

«ILBORN PROJECT ESTIMATE PROJECT HOME STAKE IENTI KAMAD SILVER CO. LAD AREA No 240 EXTENSIONS \_\_\_ ESTIMATE NO \_\_\_\_ MAINTENANCE SHOP DATE DEC 1982 LABOUR MATERIAL CCOUNT TOTAL COST DESCRIPTION YTITHAUD UNIT TOTAL RATE CODE COST COST MAINTENANCE SHOP PREFABRICATED STEEL FRAME BUILDING 50' LONG X 40'WIDE 1- PURCHASE (TAX INCLUDED) EA 1 - DELIVERY EA - ERECTION EA SITE LEVELING Sun. CONCRETE FLOOR YD3 40 FACTOR BUILT OFFICE & WAREHOUSE EA 10' LONG X ZO WIDE - PURCHASE TRANSFORT AND INSTALL (TAX INCLUDED) ELECTRICAL SERVICES - SHOP OVERHEAD FOWER SERVICE to SHOP Sum Sum LIGHTING - SHOP Sum HEATING - SHOP

WORKSHOP tools AND DENCHES

FROM

KAMAD SILVER LIMITED

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KILBORN

CLIENT: KAMAD SILVER, LTD.

PROJECT.	HONESTAKE	

PROJ No. 7560

DIV. No. 16 ESTIMATOR \_\_\_\_\_ CHECKED \_\_\_\_ SHEET No 2 OF 7

LEVEL RENABILITATION - 1750 LEVEL

DATE \_\_\_\_\_

PROJECT ESTIMATE

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KILBORN PROJECT ESTIMATE PROJECT HOMESTAKE PROJ. No 7560 CLIENT: KAMAO SILVER LAD ARFA No 250 EXTENSIONS \_\_\_\_ ESTIMATE No \_\_\_\_ REHABILITATE 2250 LEVEL DATE\_ LABOUR MATERIAL ACCOUNT QUANTITY UNIT TOTAL COST DESCRIPTION TOTAL RATE CODE COST COST COST SLASH 1040' TO 12' WIRE (4' SLASH) 38,000 413 0.70 par c.f. 0.40 ROCK BOLT ABOVE (1 pay feet) 1,040 bells 118 incl. REHABILITATE 300 VEIN H.W. DR. 2.0 MAN SHIFTS (NOMNI) + 80 SCOPHIM HOURS 80 Hours 1130 Rock helts (Allowance) 200 holk 5.23 TIMBER (Allowaxa) 12 mbf. (1000'x 12'+3') = 27 = 1400 Yn3 0300/m MISCOLLANGOUS, (FADIOSINE MAGAZINES, TOILETS, STORAGE, ETC.) - ALLOWANCE SUM

(ILBORN PROJECT ESTIMATE HOMESTAKE ENT: KAMAD SILVER CO. LTD. CLEAN OUT AUD REHABILITATE 1750 RAISE LABOUR MATERIAL CCOUNT QUANTITY UNIT TOTAL COST DESCRIPTION TOTAL RATE COST UNIT CODE COST COST COST CLEAN OUT AND REHOUS BROKEN POCK FROM RAISE - ASSUME 70% FULL COTTON

-8'110' 1417' x 60 % = 27 = 795 Yo = 100 500B	s l					$\Pi$							TI		
40 Scoops PER MAN SMITT = 10 MS ( 60 MAN HKS)				80	24.57		1920	14.14	1 1 !	110					3100
COST PER L.H.O. 110 = Free (3.91) + Tiles (1.78) + MICH (8.91) = M. 14														- -   -	
REMABILITATE AND THASER AT TOL OF RAISE (MLOW)					24.52	$\mathbf{i}$	2 75 0			1200					0150
Same, Lakeat, INSTALL THREE IN BAISE -2 HAN CREW- REHABILITATE 10 FT. ISBUT										1111		111	$\mp$		
MAN SHIFTS = 147-10:2=, 90 H.S. XR 1195				720	24.57		1.7690						山		
ROCKPULTS - 2 PER FOOT : 447 2 : 694								5.28		1720			441		43719
7/11868. = 300 bf/32" = 113 bf/ft @ 130° / subf	50					$\parallel$		430,a	21	1500			##		
BULD CHUTE - AIR OPELATED	Som			80	24.57	$\parallel$	1970	3,800		3800		111	#		572
INSTALL STEEL MILLHOLD (\$ 109/FL F.OB. SIR)	447	ft.				$\ddagger$	1200	109	43	3720			##		6620
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CLIENT: KAMAO SILVER LTO

PROJECT HONESTAKE

AREA STORE DEVELOPMENT

PROJ No 7560 DIV. No 16

AREA No. 250

ESTIMATOR\_\_\_\_\_CHECKED\_\_

\_\_\_\_\_SHEET No. 5\_\_OF\_\_

PROJECT ESTIMATE

DATE

ACCOUNT	DESCRIPTION	TYPE	QUANTITY	UNIT			BOUR					ERI	AL		CONTA	ACTO	e		TOTAL			- 1
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	SLOT RAISE - 15' 3 - 45'							$  \cdot  $		1 + 1	ment to the	$\Pi$		<u> </u>	-   -	- -			$\vdash$		4-	-
·	MANWAY RAISE -15' = 3 = 45',				_ ~-				+			+1-					111			- - -	-	-
	STOPE RAISE-220'x3 660		250			70.00/						4-1-	-  -	ļ		- -			- - <del> </del>	-+1-	++1	-
	TOTAL RAISE 750'		750	15	<u>/·</u> _	70.00/	14		5.4	500	32.00	1	2400	9	·  - -		1	-		176	150	9
	SLUSINIR STATION (10'+B'115) x3		3600	Cu Fr.	/	\$ 0.30			#	080	0.80		200	0				-		_ 3	196	0
	Sublevel Above Stopes - (418') = 48'3 = 144'								H	++		:-	+	-	-1-4	++	+	-	H		+++	-
	CROSSENT - Sublevel to Ore Par (9'11') . 170'					at .			11	TTT			THE			$\Pi$				+	111	-
	FORM 9'112' = 314		3/4	Er	_:	92.00/	7		26	890	86.40		2713	0	#	H			H	56	502	a
	DIAHONA DRILLING		400	FT				- -	++	11+		+ -	$\pm 11$		$\exists \dagger$	$\mathbb{H}$		_	H	+	06	-
	VENTILATION DOULS		5	-	200	24.47			112	1900	1000	11	500			111		1-	rtt	9	190	٦
									T	111			Thi		17				$\Box \Box \Box$			
								i												T	ITI	П
	9'+9' TRACKLESS. DRIETS.								1 1											TI		
	71/2 MIN SHE'S @ 28.78/HI = 61.40/F4							1.	11			1								П		
	A 00 20% FOR HIXAN, 12.28								11			LL				Li						
	TOTAL LABOUR . 73.68, A.																		Ш			
	Non HAT = INCL. L. H.O. COSTS: 0.80/11 = 72.80/1								11			$\perp$					Lil					
									- -							11				$\perp$	Ш	
i	RAISING (5'18')							_	.!_	L		11					Ш		ĿL		Ш	
	71/11/2 MAN SHIETS NT 2879/11 = 61.40/F								11	111		4-1-							Ш	$\perp \! \! \perp$	Ш	
	100 15 % FOR MUXING : 8.60/6								11			1.1.			_ _ _	Ш				11		
	TOTAL LABOUR: 70.00								11	1.1.1			-1-1-1-							11		
	Ann MAY - INCL. MUCKING CUST 0.80/11: 32.00								-	11:		-									Ш	
												- - -	-44		$- \bot$	Ш	111	-		11		
	9'x12' SUB-LEVEL - TRAKLUSS						:		+	111		11-	414		1	11-1	111		44	44	111	Ц
	9 1/3 MANSHIETS 28.78/18 = 76.75							1	11			1	111		-11	111	111	_		1	Ш	
	100 20% FOR MUCKING 15.26								- -	1-1-		1-1	-			-   -	1 -1	-			1.1.	
	Torthe Labout 97.00								1	111	1	11	111		11	111	111	1	11	11		1

CLIENT: KAMAD SILVER LD

PROJECT ESTIMATE PROJECT HOMESTAKE AREA MINE EQUIPMENT, SUPERVISION PURME AREA NO 250

DATE \_\_\_

CCOUNT	DESCRIPTION	TYPE	QUANTITY	UNIT			BOUR			3 L	EF	RIAL						TOTAL		
CODE	DESCRIPTION	TTPE	QUANTITY	UNIT	UNIT	MH	MH	- (	COST	COST		COST	COST		cos	T	7	TOTAL		351
								П			i				$\prod$			Ш	T	TT
	SLUSHER HORTS SOME ELECTRIC (USEO)		4					П	100	9,000		3600	0		TT				3	200
	SCRAPEIS FOR ABOVE .		4						20	3.500		19000							1	23
	BLOCKS FUR ABOVE		20				l		IIIH	- 100	Ĺ	200	0							200
	ELECTRICS FOR ABOUT		4		19 10 200 00 00 00				100	2,000		500					_			200
	13 TON MINE TRUCKS - DYEKHAHL		2						300	10,xx		2000	d	-	-!		-		2	300
	22403 L.H.D. UNITS		3					#	950	10,000		3000	٠				-		34	
	CHECK ROCK DRILLS AND OTHER PLUMS										•				,				H	
	EXCIPMENT EXON DEKALB		LOT					П	1000		-	1700	<i>a</i>	II	į.	П		Ш	1	20
	TRANSCORTATION							$\parallel$	1 200	2	•	200			1		_		1	0
	SUB TOTAL - EQUIRALINT								21700	<b>X</b>	:	12600		1	<u>-</u>		-		4 7	2 7
	SUPERVISION DURING DEVELOPHENT (6 Mos)			ļ					157500					1	+		$\perp$	- - -	5	15
	POSITION ANNUAL SALAKY							#			-			#	#		1		$\sharp$	-
	MINE SUP'T (1) # 50,000							+	tilli		- <b>i</b> -			1	+	H	+	H	$\dagger \dagger$	+
	SHIFT BOSS (2) 36100.2 = 76,000							$\Pi$			Ţ				Hi	$\Pi$			II	Ī
	GEOLOGIST 40,000										1									
	SURVEYOR 36,000					51						11								
	SURVEY ITELACE ZADDO																			
															T					
	Tome \$ 24,000							П												
	+26.5% FEMILE 66,000								444		-		-	-	1		- -	Ш	#	-
	TUTAL PER YEM = 315,000							#	<del>††  </del>		1			+	++	Hi				
	FOR 6 NO-1015 : 157,500							1	!		_			1	1				14	14
	Marine de la compansa							-	11-1-1		-	11-1-1				-1-1			1	

KILBORN
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X     L  B ( O )  B   N					PROJECT	ESTIMATE	
	PROJECT	PROJ No 7560			TYPE		_
LIENT:		DIV No	ESTIMATOR	CHECKED	SHEET No	_7_or7	_
	AREA	AREA No _ 250	EXTENSIONS	APPROVED	ESTIMATE No	·	_
	OVERHEND COSTS - SIX MONTH PERI	OR			DATE		

ACCOUNT	0.5.0.0.0.7.0.0	TYPE	QUANTITY	UNIT		L	BOUR						RIAL		OTH	E.E.				COST	
CODE	DESCRIPTION	ITPE	QUANTITY	וואט	UNIT	TOTAL	RATE		COS	s T	CO		COST		UNIT	co	S T	101	AL (	2031	
								П								$\Pi\Pi$	TTT	$\Box \Box$	П	Ш	Т
	ADMINISTRATION.							T								$\Pi\Pi$			T	ПТ	П
	MANAGERS SALARY - 6,300 x 6 1105.								3	780	c								] 3	78	p
	CONTRACTED FUNCTIONS -3.800 × 6 HOS.															12	2800			28	0
	CONSULTING SERVICES 3500 x 6 MOS												1111		1 5 0	2	1000		Z	10	
	COMMUNICATIONS 1,900 x 6 HOS		0.5													1	1 1 1 1		1/	10	0
	FREIGHT COSTS 1,200 x 6 HOS					. 1							72	00					П	72	٥
	TRAVEL COSTS 1,000 x 6 HOS													111			6000		$\Pi$	50	0
	RECEVITHENT 1900 Y 6 Mos																1200		1/	110	
	INSURANCE 7,500 × 6 1105								11					!!			5000		14	500	9
									3:7	780	0		1.172	100	=	111	7600		16	26	0
-	TOTAL ADMINISTRATION								11				111	11		1 1					
									11	11				111_		1			$\coprod$	Ш	
	OPERATING SUPPLIES (BALANCE INCL. IN UNIT)							.[.]	-11				. l i . l .			!					
	DIESEL OIL - COMPRESSOR & GENERATED		26,000	GAL		•					1.	20	442	00							
	PROPANE		21,200	GAL					. [_]_			14	242						Ш		
	MISCELLANEOUS SUPPLIES. (ESTIMATE)							$\perp$	-	Ш			200			1					
								.11	11				TLi.I.	<u> </u>		1 1	_[_[_]	_     _	11		
	TOTAL OPERATING SUPPLIES							$\perp$	<u>i</u> i				887	00		1 11			13	840	0
			11 0					1.			1.										
	MAINTENANCE MATERIALS				1																
	ELECTR. GENERATOLS. \$1,400 +6 HOS												84	100					П	E11	0
	AIR COMPRESSORS. 1.200 +6 Mos							i					72	00	_ = [		TTT		П	72	٥
	Punps. 200 x 6 11x													00						12	0
	BUILDINGS. 400 x61105													100					П	24	0
11	SMALL VEHICLES . 1.100 x 8.164							i	. : i				166	00						6.6	D
	ELECTRICAL MISCELLANEWS, BOOKENING								11					100			TITI		T	19	P
	MECHANICAL MISCELLANEOUS. 600 AGINA							il	11				at	300					IT	18	0
													1111						П		
	TOTAL MAINTENANCE MATERIALS								1				354	100					3	54	ol
									H				Tiil					IT	IT		T
	And the second s							1	1							1-1-1			IT	FIT	
								T	T	1			1111	H		TITI	1111		TT	111	-1
		-						11	11	1-1-	-		THE			T1-1-1			1-1-	111	1

CLIENT: KAMAD SILVER CO. LAD

PROJECT ESTIMATE

PROJECT HOMESTAICE PROJECT FOR TYPE

DIV NO 16 ESTIMATOR CHECKED SHEET NO 1 OF 6

AREA CONCENTRATOR AREA NO 30 EXTENSIONS APPROVED ESTIMATE NO 10 CONCENTRATE NO 10 CONCENTRAT

DATE DEC 1982

ACCOUNT	0.5.0.0.0.7.10.11	TYPE	QUANTITY				A B O U R		MATERIAL			*****
CODE	DESCRIPTION	TYPE	UJANIITY	UNIT	UNIT	TOTAL	RATE	COST	COST COST	UNIT	COST	TOTAL COST
												$\Box\Box\Box\Box$
	MODIFICATIONS to EXISTING DEKA	415	M.LL						i i i			
										11.		
									[[[]]	!		
	SULPHIDE CIRCUIT							.   .   .   .   .   .   .   .		111		
										<u> </u>	1111111	
				11.000		new conse				1.1		
_!	INSTALL EXISTING 15" DIA CYCLONE			2				18.0	: :	<u> </u>	!   i !	180
	- NEW Supporct steel			₩.				1 1 400		2. 0:0		3600
								111111			1-11-	
				ļ							1-1-1-1	
2	INSTALL EXISTING CYCLONE FEED PUMP							<del>                                      </del>		! ! !	1-11-11-1	
	AND MOTOR - STAND BY SERVICE			No				3 100		11	1-1-1-1-1	210
	- NEW CONCRETE PUMP FOUNDATION			-No.				114:0:0		.00	1-1-1-1-1	300
					·	ļ					1-1-1-1-1-1-1	
								- - - - - -		<del></del>	1-1-1-1-1	
3	MODIFY EXISTING GRINDING CIRCUIT		Sum							· · · · · · · · · · · · · · · · · · ·	- <del> </del>  +-	
	PIPING VALVES AND FIFTINGS - STAND-B							++++-1-+1-		<del>! :  </del>	1-1	
	PUMP AND CYCLONE		Sum.					9500		000	1-1	1550
								┼┼┼┼┼		HI	1 + 1 + 1 + 1 + 1 + 1 + 1 + 1 + 1 + 1 +	
								- <del> - - - - - - -</del>  -		111	1+1++++	
4	Modify Existing Flotation circuit		Sym					++++++		00	<del></del>	
	PIPING VALVES AND FIFTINGS							11500		0.0	1-1-1-1-1-1	1 21 30
								- <del> - - - - -</del>  -		+++	1+++++	
5	THICKENER TANK - 14'DIA X8' MS							-i-l-t-t-t-t-		i- -	1+1+1+1	
	C/w WEIR - NO LINING			N.			-	1 200			<del>11      </del>	1790
	- Support Steel		Sum						:/ 3			
	- Concrete Fourantions		Sum					4200		1.0.0	1	850
								++-755		11		
										11		
۷	THICKENER MECHANISM C/W DRIVE AND									11		
	1.5 H.P. Motor - 14' DIA		F 1 1	No				4900	11.8	50.0		2340
								1111111				

CLIENT: KAMAD SILVER CO. LA

				PROJECT	ESTIMATE
Homestrice	PROJ No 7560			TYPE	
	DIV. No	ESTIMATOR	CHECKED	SHEET NO	3_0F_6
CONCENTRATOR	AREA No	EXTENSIONS	APPROVED	ESTIMATE N	·
		DIV. No	DIV. No	DIV. No	Ho~E5 trikE PROJ No 7560 TYPE

ACCOUNT	DESCRIPTION	TYPE	QUANTITY		UNIT TINU		LABOUR					TER	IAL								TO	TA.	cos	C T	
CODE	DESCRIPTION	ITTE	QUANTIT	ONT	UNIT M H	TOTAL	RATE		СО	s T		COST		cos	5 T	UN CO	ST		co	s T			IAL	CUS	**
	(4) 2										$\prod$				$\prod$						Π				I
	SULPHIDE CIRCUIT - CONTINUED															1		1						Ш	
													1					11	!!						
	CONCRETE FOUNDATIONS FOR 1-town					1 5 0 0 000		1			1			1-1	H	i		11	-: 1	Ш	11				
	+~ b +wo pumps		- Sum							280	ابت		1	1.3	3 1 9			Ш			1.			_ 5	90
								1.	!!!				1	ļ.,	11			Ц	-14	11	1	_			1
	# 1907 1 WWW 11 WW 20 W 20 W 20 W 20 W 20 W 20										!							i.			11.				
15	MUDIFY EXISTING THICKENING CIRCUIT							_	1 1					1.		1				11	1			1	
	PIPING VALVES AND FIFTINGS		Sum		1713			1.		8 5	010		1:	1 3	3 50	, ,		L						12	00
	- NEW PIPING VALVES AND FIFTINGS		Sum					1.	1	2 5	0:0				6 5	טיפ				11	1	1		19	طو
											<u></u>				1			+	;	11	4-1-				1
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									-,-!		-11				-	<u>i                                    </u>		1	1 1	-   -	1	$\perp$		Ш	1
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	BAKITE CIRCUIT							-ļ-		4	+1.			·	1.					11	- -	4	-		-
								1.	. [ ]		1.				-	÷		1-		-14	-11-	_ -	4-		-
									::		-				1	-		-1	11	-44	4	- -	1	+	-
	Install Existing REGRIND MILL							-	<del>- ;- ;</del>	$\perp$					-11	!		1		44	11-	4	_ _		_
	d'DIA × 6' LONG			_~.				1	1	7 8	00		1	+ 1	+1	-		4-1	11	4	#	_ -	-	_ 7	80
	- NEW Motor 100 HP		ļ	N.				1		13	00			5				Ц		-1-1	41	_ -		11	00
	- NEW DRIVE BOLTS, GUARD Etc		i	No				-	14	111	00			3				4.1	L		1.1_	_		4	70
	- Concrete Foundation		5 vm					-	:	49	90		1	6	810	10		1	-11	4	11			11	70
							ļ	-	! !				1	!!	1	11		11	$\perp$	Ц.	11-	_	_	+	-
				-				+	1	1	$\perp$		14	1	1	1		11	41	$\perp$	41	4	1	H	1
2	FEED AND DISCHARGE CHUTES		_ 2	~·					L,	2 !	olo		- j	1	3 81	20		11	44	-11	11	_		_ 5	90
		ļ	ļ						. <u>L</u> .	_   -					- - -				-;	-1-1		_		Ш	
								-	1	- -			ļ.,	<u> </u>				1-1	4	-1-1-					+
3	Pump Box - REGRIND MILL - NO LINING		!	No.				+		1 3	olo				44	00	-	+-	- -	+1	+-			6	00
								- -			+1		1		-1-1	-		+-		- -	+-	-	-	+	
·								-	1	H	4				1			H	11	11	1	-			+
	Pump - RECKIND CYCLONE FEED C/W									-	+1		1.		11			1-1	+	-1-1	1-			-	
	75 HP Motors		I	N.			1	-		18	00			11 :	1 10	w		1	-	.1.	11.		1.	6	20

CLIENT: KAMAD SILVER COLTO

AREA CONCENTRATOR AREA NO 300

PROJECT ESTIMATE

ESTIMATOR \_\_\_\_\_ CHECKED \_\_\_\_\_ SHEET No 4 OF 6

DATE DEC 1982

						ABOUR		MATERIAL				٠.												
DESCRIPTION	TYPE	QUANTITY	וואט	UNIT	TOTAL MH	RATE		COS	s T	out on	COST		COS	T	COST		co	<b>9 T</b>		10	DTAL		31	
									П			П	T			Ti	П		П		$\Pi$	П	П	
BARITE CIRCUIT - CONTINUED																							П	
														Ш		11								
CYCLONE - 10" DIA REGRIND ofw OVER													11			11.	1!		1.1				11	
		2	~.						3 7	هِ هِ		11	111	50	b	11	L.		Ш			15	20	2
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								11				11	1!	Ш		-	1	Ц.				.	14	1
CONCRETE FOUNDATIONS FOR PUMP							<u>.</u>	11		4			-	14		11.	·	-	Ш		11	H		1
Box AND PUMP	-		No.				4		21	010			1.1	40	o	<u> </u> :	ļ	-	14	_	11	13	50	2
								-11	$\perp$	4		<u> </u>	1:	4	_	ļ	<u> </u> :	-	H	_	1	H	Н	1
									-1-1			41		1	-	-	1	1	H	_	H	H	4	4
							4	4	$\perp$	4			++			<u> </u>	1	1	1		11	$\vdash$	1.1	4
		!	N.						₹ !	9 9.		41	++	++-		-	H		$\sqcup$		+	3	10	2 5
							4	-11	11	00		+1				-	1	- -		_	₩	11	80	3 6
	-							-1-1	16	0.0		1				1		-		.	ļ. J.	12	80	c
- NEW MOTORS AND ACITATOR MECHANISMS		Sum				-		-	5 5	00		+	18	00	0			+	H	-	╁	23	50	k
Relocate Existing Fump Boxes		5	No.					1	96	00		İ	Ħ-	1			I		+			9	6	
			-					11	$\perp$	+		4	11			1	1	4	Ш	_	4	-	H	+
Puna 51 tata sur to che Davis	-						+	+	+	H		+	+	H		1	H		+	-		$\vdash$	H	+
	·	<b>*</b> —	1				+1	十寸	1	1		++	1-1-	1		1	H		H		$\dagger \dagger$	20	1	$^{\dagger}$
MAD MAJOK - NEW	-		1.00			-	11		1 2	i		++	+177	70	-	1	1	-	H		+	12 0	115	1
													扛				士						ഥ	1
FOUNDATIONS FOR PUMP ISONES AND								11		1						Ш			Ш				Ш	1
NEW PUMPS		10	~6.				Ш	11	07	00		11	13	30	0	1	1		Ш		Ш	21	00	2
								- - -	Н					- -	-	1	1	1	1	-		-	44	4
Post des de la	-						+-	++		+-		++	-			+	+-	-	++	$\vdash$		H,		+
FOM P 120X - BARITE THILINGS	-		~o.	1		<u> </u>	-	$\forall$	16	00		++	廾	120		ΙŤ	Ħ	$\vdash$	H		+	-16	1	Ť
								$\Box$				I	I	Ш										
Pump - BARITE TAILINGS C/W DAINE																11	L							1
CIMB WO for 4. " 3.	-		No						18	op			. 4	60	0	11	1				. .	6	40	٨
	Pump 160x - Enrite trilings Pump - BANITE TAILINGS C/W DAINE	CYCLONE - 10" DIA KEGKIND C/W OVER  AND UNDERFLOW LAUNDERS  CONCRETE FOUNDATIONS FOR PUMP  BOX AND PUMP  Flotation cell - Dismantle and  Relocate Existing c/w Drive  - Launder Modification  Support steel  - Now motors and agentor mechanisms  Relocate Existing pump Boxes  Pump - Flotation circuit - c/w Drive  AND motors - NEW  Foundations FOR PUMP BOXES AND  NEW pumps  Pump Box - tancite tailings  Pump Box - tancite tailings  Pump Box - tancite tailings	Crelone - 10" DIA KEGRIND OFW OVER  AND UNDERFLOW LAUNDERS 2  CONSIDER FOUNDATIONS FOR PUMP  BOX AND PUMP  FLOTATION CELL - DISMANTLE AND  Relocate Existing ofw DRIVE  - LAUNDER MODIFICATION  - Support Steel  - NEW MOTORS AND AGITATOR MECHANISMS  Relocate Existing pump BOXES  FOUNDATION CIRCUIT - Ofw DRIVE  THE MODIFICATION CIRCUIT - Ofw DRIVE  THE MODIFICATION CIRCUIT - Ofw DRIVE  TO MODIFICATE FOR PUMP BOXES AND  NEW PUMPS  10  PUMP BOX - EARLIE TAILINGS OFW DRIVE	Cyclone - 10" DIA REGRIND C/W OVER  AND UNDERFLOW LAUNDERS	BARITE CIRCUIT - CONTINUED  CICLONE - 10" DIA KEGRIND CON OVER  AND UNDERFLOW LAUNDERS.  CONCRETE FOUNDATIONS FOR PUMP  GON AND PUMP  FLOTATION CELL - DISMANTLE AND  RELOCATE EXISTING CON DRIVE  - LAUNDER MODIFICATION  - Support Steel  NSW Motors AND REITHON MECHANISMS  RELOCATE EXISTING PUMP BOXES  TO NO.  PUMP - FLOTATION CIRCUIT - CON DRIVE  AND MOTORS FOR PUMP BOXES  FOUNDATIONS FOR PUMP BOXES AND  NEW PUMPS  10 No.  PUMP BOX - TENNITE TAILINGS CON DRIVE	BARTE CIRCUIT - CONTINUED  CICLONE - 10" DIA KEGRIND OFW OVER  AND UNDERFLOW LAUNDERS . 2 No.  CONSERTE FOUNDATIONS FOR PUMP  CON AND PUMP 2 NO.  Flotation cell - Dismantle and Relocate Existing of Daive 1 No.  Launder Modification of Modifier Sum  Supposed Steel Sum  No Motors and Relation mechanisms Sum  Relocate Existing pump costs 5 No.  Pump - Flotation circuit - chu daive 5 No.  Foundations for pump makes and No.  Foundations for pump makes and No.  Pump 20x - tracite tailings 1 No.  Pump 20x - tracite tailings 1 No.	Braite Circuit - Continued  Ciclone - 10" Bia Regard of over  and unduplow launders  Concrete found Athens For Pump  Box and pump  Example Resisting of Daire  - Launder Madified Mechanisms  Support Steel  - Now motors and agitator mechanisms  Relocate Existing pump Boses  Florate Existing pump Boses  Foundations for Pump Boses  The No.  Pump - Rotation Circuit - And Daire  The pumps  10 No.  Pump Box - Energy tallings of Daire  The pump Box - Energy tallings of Daire  The pump Box - Energy tallings of Daire  The pump Box - Energy tallings of Daire	BARITE CIRCUIT - CONTINUED  CYCLORE - 10" DIA REGRIND CIN OVER  AND UNDERFLOW LAUNDERS.  CONSISTE FOUNDATIONS FOR PUMP  ZON AND PUMP  RELOCATE EXISTING CIN DRIVE  - LAUNDER MODIFICATION  - Supposed steel  Now Motions and Activities accumulante  Sum  Relocate Existing pump coxes  5 No  Pump - Flotation CIRCUIT - C/N DRIVE  TOURDATIONS FOR PUMP MONES AND  NEW PUMPS  FOUNDATIONS FOR PUMP MONES AND  NEW PUMPS  10 No  Pump 120x - travite failings  The Days 10 No  Pump 120x - travite failings  The Days 10x - travite failings  Th	BARITE CIRCUIT - CONTINUED  CYCLORE - 10 DIA REGRIND COVER  AND UNDERFLOW LAUNDERS.  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		DIV No16	ESTIMATOR	CHECKED	SHEET No _ 5 OF 6
AREA	CONCENTRATOR	AREA NO 300	EXTENSIONS	APPROVED	ESTIMATE No
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PROJECT ESTIMATE

DIV. No. 16 ESTIMATOR CHECKED SHEET No 6 OF 6 .

AREA NO 300 EXTENSIONS APPROVED ESTIMATE No

DATE DEC 1982

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12.0 OPERATING COST

#### 12.0 OPERATING COST

#### 12.1 SUMMARY

The projected operating cost per short ton is \$67.92. The cost does not include shipping of concentrate or smelter charges which are considered in the net smelter return calculation (Section 8.0).

The operating cost is summarized as follows:

Description	Monthly Cost	Cost per Ton
Administration (Including Ore Haulage)	\$ 112,800	\$ 13.43
Mining	258,400	30.75
Milling	147,400	17.55
Contingency on Operations (10%)	51,900	6.19
TOTAL OPERATING COST	\$ 570,500 ======	\$ 67.92

### 12.2 BASIS OF ESTIMATE

The estimate is based on mining, transportation, and milling methods described elsewhere in this Report.

#### 12.2.1 Labour Rates

Labour rates are based on a Labour Contract currently in force at a nearby mining operation. The rates have been modified to allow for experience from recent operations at the site and at other similar operations.

Fringe benefits included are outlined below in Table 12-1.

#### TABLE 12-1

#### PAYROLL BURDEN

<u>Item</u>	Percent
Unemployment Insurance	1.1
Canada Pension Plan	1.0
Workers' Compensation	8.6
Pension and Life Insurance	4.0
Vacation with Pay	4.9
Statutory Holidays	4.6
Unscheduled Overtime	2.3
TOTAL PAYROLL BURDEN	26.5 ====

#### 12.2.2 <u>Material Costs</u>

Material costs are based on budget quotations received from suppliers and are valid for December, 1982.

### 12.2.3 <u>Transportation Costs</u>

Transportation costs are based on budget quotations and are valid for December, 1982.

# 12.2.4 <u>Electrical Energy Costs</u>

Electrical energy costs are based on British Columbia Hydro costs at the mill site and on diesel power generation at the mine site. Fuel costs for diesel power generation are valid for December, 1982.

# 12.3 ADMINISTRATION COSTS

Administration costs are summarized as follows:

	Item	Monthly Cost
	Mine Manager's Salary	\$ 6,300
	Contracted Functions	3,800
	Consulting Services	3,500
	Communications	1,900
	Ore Haulage	84,000
	Material Freight Charges	1,200
	Travel Costs	1,000
	Road Maintenance	1,700
	Recruitment	1,900
	Insurance	7,500
	TOTAL ADMINISTRATION COSTS	\$ 112,800 ======
	Cost per Ton - \$ 13.43	
12.3.1	Mine Manager's Salary	
	Item	Annual Cost
	Base Salary	\$ 60,000
	Payroll Burden at 26.5 Percent	15,900
	TOTAL ANNUAL COST	\$ 75,900
	Total Monthly Cost = \$ 6,300	
12.3.2	Contracted Functions	en e
	<u>Item</u>	Annual Cost
	Accounting - \$600.00 per Week	\$ 31,200
	Secretarial - \$200.00 per Week	10,400
	Watchman - \$300.00 per Month	3,600
	TOTAL ANNUAL COST	\$ 45,200
	Total Monthly Cost = \$ 3,800	

## 12.3.3 <u>Consulting Services</u>

<u>Item</u>	Annu	al Cost
Mining Engineering (5 Days per Month at \$250.00 per Day)	\$	15,000
Metallurgical Consulting (2 Days per Month at \$450.00 per Day)		10,800
Environmental Consulting (3 Days per Month at \$450.00 per Day)		16,200
TOTAL ANNUAL COST	\$ ===:	42,000

### 12.3.4 <u>Communications</u>

Total Monthly Cost = \$ 3,500

<u>Item</u>	Mon	thly Cost
Telephone Rental at the Mine Site	\$	100
Telephone Rental at the Mill Site		50
Telephone Rental at the Manager's Home		50
Long Distance Calls		500
Telex Rental		900
Telex Charges		200
Courier Services		100
TOTAL	\$	1,900
TOTAL ANNUAL COST (\$1900 x 12)	\$ ==	22,800
Total Monthly Cost = \$ 1,900		

## 12.3.5 Ore Haulage

Ore haulage from the mine to the mill costs \$10.00 per ton.

TOTAL ANNUAL COST \$ 1,008,000

Total Monthly Costs = \$ 84,000

### 12.3.6 <u>Material Freight Costs</u>

Delivery costs of supplies and services are estimated to be \$1200 per Month.

TOTAL ANNUAL COST (\$1200 x 12)

14,400

Total Monthly Cost = \$ 1,200

Total Monthly Cost = \$ 1,000

### 12.3.7 <u>Travel Costs</u>

<u>Item</u>	Annua	1 Cost
Manager's Automobile		
Rental - $$320.00 \times 12 \text{ Months} = $3,840$		
20,000 Miles at \$0.22 per Mile = 4,400	\$	8,240
Air Travel		
(15 Return Flights Vancouver to Kamloops)		
\$151.20 x 15		2,260
Travel (Hotel and Expenses)		
Estimate 15 Days at \$100.00		1,500
TOTAL ANNUAL COST	\$	12,000

### 12.3.8 Road Maintenance

A local contractor has estimated that the following costs would be involved to maintain 8 miles of road at the mill site:

	==	=======
TOTAL ANNUAL COST	\$	20,000
Crushed Gravel		2,700
152 Hours at \$48.00		7,296
192 Hours at \$52.00	\$	9,984

Total Monthly Cost = \$ 1,700

#### 12.3.9 Recruitment

<u>Assumptions</u>	<u>Annual Cost</u>
One Staff Move per Year	\$ 10,000
One Recruiting Fee per Year	2,500
Ten Hourly Recruits per Year	10,000
TOTAL ANNUAL COST	\$ 22,500 ======

Total Monthly Cost = \$ 1,900

#### 12.3.10 Insurance

A value of \$6,000,000 was estimated for those items that might be destroyed by fire. The insurance premium was placed at 1.5 percent per year of the insured value.

TOTAL ANNUAL COST - \$ 90,000

Total Monthly Cost = \$ 7,500

### 12.4 MINING COSTS

Mining costs include all operations at the mine site. These costs are as follows:

Item	Monthly Cost	Cost per Ton
Supervision and Technical Salaries	\$ 26,200	\$ 3.12
Mine Production Crew Salaries	130,600	15.54
Maintenance and Surface Operations		
Crew Salaries	19,100	2.27
Mine Operating Supplies	63,700	7.58
Mine Maintenance Materials	18,800	2.24
TOTAL MONTHLY COST	\$ 258,400	-
TOTAL COCT DED TON		<b>^</b> 20 77
TOTAL COST PER TON	-	\$ 30.77

## 12.4.1 Supervision and Technical Salaries

<u>Position</u>	Number	Salary	Annual Cost
Mine Superintendent	1	\$ 50,000	\$ 50,000
Shift Boss	2	38,000	76,000
Geologist	1	40,000	40,000
Surveyor	1	35,000	35,000
Survey Helper	1	24,000	24,000
Clerk	1_	24,000	24,000
Subtotal	7		\$ 249,000
Payroll Burden at 26.5 Percer	nt		66,000
TOTAL ANNUAL COST			\$ 315,000

Total Monthly Cost = \$ 26,200 Cost per Ton Mined = \$ 3.12

### 12.4.2 Mine Production Crew Salaries

Position	Number	Hourly Rate	Annual Salary	Annual Bonus	(%) Bonus	Annual Cost
Development Miner	4	\$ 13.00	\$ 27,040	\$ 20,280	(75%) \$	189,2
Stope Miner	16	13.00	27,040	13,520	(50%)	648,9
L.H.D. Operator	4_	12.50	26,000	13,000	(50%)	156,0
Truck Driver (U/G)	2	12.50	26,000	7,800	(30%)	67,6
Spare Miner .	2	13.00	27,040	13,520	(50%)	81,1
Timberman	1	14.00	29,120	9,610	(33%)	38,7
Labourer	2	11.00	22,880	5,720	(25%)	57,2
Subtotal	31				\$	1,238,8
Payroll Burden at 2	6.5 Percen	t			•	328,3
TOTAL ANNUAL COST					\$ =	1,567,2

Total Monthly Cost = \$ 130,600 Cost Per Ton Mined = \$ 15.54

## 12.4.3 <u>Maintenance and Surface Operations Crew Salaries</u>

Position	Number	Hourly <u>Rate</u>	Annual <u>Salary</u>	Annual Cost
Mechanics Leader	1	\$ 18.00	\$ 37,400	\$ 37,400
Mechanics	3	14.00	29,120	87,360
Electrician	1	14.00	29,120	29,120
Labourer - Changeho	use 1	13.00	27,040	27,040
Subtotal	6			\$ 180,920
Payroll Burden at 2	6.5 Perce	nt		47,980
TOTAL ANNUAL COST				\$ 228,900 ======

Total Monthly Cost = \$ 19,100 Cost per Ton Mined = \$ 2.27

# 12.4.4 <u>Mine Operating Supplies</u> (Summary - Annual Cost)

<u>Item</u>	Annual Cost
Drill Steel and Bits	\$ 35,000
Explosives	219,300
Timber	50,000
Rock Bolts	47,600
Diesel Oil Costs -	146,900
Propane - Mine Air Heating	49,300
Tires	15,700
Diamond Drilling	69,000
Pipe	32,900
Small Tools	38,900
General Consumables	60,000
TOTAL ANNUAL COST	\$ 764,600
Total Monthly Cost = \$ 63,700	

Total Monthly Cost = \$ 63,700 Cost per Ton Mined = \$ 7.58

#### 12.4.5 Drill Steel and Bits

Cost per Foot Drilled, Bits (1-3/8" Diameter Bits = \$25.25 : 375 Feet)	\$ 0.067
Cost per Foot Drilled, Drill Steel (\$38.55 ÷ 900 Feet)	0.043
TOTAL COST PER FOOT DRILLED	\$ 0.11 =====
Annual Footage Drilled:	
(a) Drift and Subdrift (2000' x 37 Holes)	74,000 Feet
(b) Raise (2400' x 23 Holes)	55,000 Feet
(c) Stoping (90,000 t/y at 0.60 t/f Drilled)	149,000 Feet
(d) Rock Bolts (5600 Bolts at 6')	34,000 Feet
(e) Miscellaneous (Add 2% to Above)	6,000 Feet
TOTAL	318,000 Feet
TOTAL ANNUAL COST (\$0.11 x 318,000 Feet)	\$ 35,000 ======

#### 12.4.6 Explosives

(a) Thirty-four (34) 10 foot holes blasted per round (advance equals 9.5 ft)

Footage loaded per round (34 holes x 8.5 ft) = 289 Feet Explosive cost (water gel) (\$318.55 per 221 lb) = \$1.44 per lb Cost per round (\$1.44 x 289 ft x 0.47 lb per ft) = \$195.60

Detonators (Nonelectric) - \$1.81 Each Primers - 0.44 Each Subtotal \$2.25

Subtotal \$ 2.25

Cost (34 holes x \$2.25) = \$ 76.50

TOTAL COST PER ROUND = \$ 272.10

TOTAL COST PER YEAR (2000 ft x \$272.10 ÷ 9.5) = \$ 57,300

#### (b) Raise

Twenty (20) 8 foot holes blasted per round (advance 7.5 ft)

Footage loaded per round (20 holes x 6.5 ft) = 130 Feet Explosive cost per round (1.44 x 130 x 0.47) = \$87.98 Detonators and primers (20 holes x 2.25) = \$45.00 TOTAL COST PER ROUND = \$132.98 TOTAL COST PER YEAR (2400 ft x 132.98  $\div$  7.5) = \$42,600 =======

#### (c) Stoping

TOTAL ANNUAL COST

Average explosive factor equals 0.75 pound per ton broken. Use 50 percent water gel and 50 percent AN-FO.

Water Gel Cost \$ 1.44 per 1b AN-FO Cost \$ 61.87 per 221 lb (\$0.28 per lb)  $(\$1.44 + \$0.28) \div 2 = \$0.86$  per 1b Average Cost Explosive Cost 90,000 tons mined x 0.75 lb per ton x \$0.86equals \$58,100 Detonators and Primers (6.2 tons/hole)  $$2.25 \times 90,000 \div 6.2 = $32,700$ TOTAL STOPING \$ 90,800 Drift Cost 57,300 Raise Cost 42,600 TOTAL \$ 190,700 Add 15% for secondary blasting, detonating cord, electric caps, loading sticks, magazine rental, etc. 28,600

\$ 219,300

### 12.4.7 <u>Timber</u>

A cost of \$50,000 has been allowed for timber.

### 12.4.8 Rock Bolts and Accessories

Rock bolts estimated are:

Drifts and Subdrifts (2000 ft at one bolt per 2 ft)	<b>-</b> .	1000 Bolts
Raises (2400 ft at one bolt per 1.5 ft)	-	1600 Bolts
Stoping (Bolting on a 4 ft x 4 ft pattern) (Allow 30 tons per bolt)		
Bolts Required (90,000 ÷ 30)	-	3000 Bolts
TOTAL		5600 Bolts

#### Annual Cost:

5,600 5/8 inch x 6 foot rock bolts at \$3.77	-	\$ 21,100
6,000 1-3/8 inch diameter shells at \$1.18	-	7,100
6,000 $1/4$ inch x 4 inch x 4 inch washers at \$0.52	-	3,100
4,000 1/8 inch x 4 inch x 5 foot straps at \$3.29	-	13,200
14,000 square feet of wire mesh at \$0.22	-	3,100
TOTAL ANNUAL COST	_	\$ 47,600

### 12.4.9 <u>Diesel Oil Costs</u>

Diesel oil costs are calculated as follows:

- (a) J.D.T.-413 13-ton trucks operate 10 hours per day for 250 days at 3.2 gallons per hour 8,000 gal
- (b) Two cubic yard LHD units each operate 12 hours
  per day for 250 days at 2.3 gallons per hour 13,800 gal

(c)	Air compressors operate 18 hours per day for 250 days at 4.4 gallons per hour	-	19,800 gal
(d)	Electric generator operates 24 hours per day for 250 days at 4.6 gallons per hour and 115 days at 1.7 gallons per hour	-	32,300 gal
(e)	Miscellaneous mobile equipment, (loader, undergound service truck, bulldozer, etc.); allow 50 gallons per day for 250 days	-	12,500 gal
	TOTAL ANNUAL FUEL CONSUMPTION	-	86,400 gal
	TOTAL ANNUAL COST (86,400 gallons at \$1.70 per gallon)	-	\$ 146,900 ======

# 12.4.10 Propane Costs

Mine air heating will be required for 4 months per year for an average air temperature rise of  $20^{\circ}F$ . The annual propane consumption will be:

60,000 cfm x 4.3 months x 25 days x 24 hours x 0.075 lb per ft x 0.24 Btu per lb per deg	x 60 gree F	minutes x 20°F	
110,000 Btu per gallon			
ANNUAL CONSUMPTION	-	31,000	Gallons
Mine dry heating, including hot water will consume an average of 540 gallons per month			
for 12 months	-	6,500	Gallons
The maintenance shop will require	-	4,000	Gallons
The residence will require	-	800	Gallons
TOTAL ANNUAL CONSUMPTION	-	42,300 =====	Gallons
Cost at \$1.14 per gallon	-	\$ 48,200	
Tank rentals (\$544.00 x 2)	-	1,100	
TOTAL ANNUAL COST	-	\$ 49,300	

======

#### 12.4.11 <u>Tires</u>

The tire life on the 13-ton mine truck will average 2700 hours. The annual cost:

 $($850 \times 4 \times 250 \text{ days} \times 10 \text{ hours}) \div 2700$  - \$ 3,200

Two cubic yard LHD units will average 1800 hours per tire. The annual cost to operate 2 units is:

 $(\$790 \times 4 \times 2 \times 250 \text{ days } \times 12 \text{ hours}) \div 1800 - \$10,500$ 

Other mobile equipment (allowance) - \$ 2,000

TOTAL ANNUAL COST \$ 15,700

#### 12.4.12 Diamond Drilling

Allow 3000 feet of contract diamond drilling per

year at an overall cost of \$18.00 per foot - \$54,000

Allow a further 1500 feet of short hole drilling by mine crews at \$10.00 per foot

15,000

TOTAL ANNUAL COST

\$ 69,000

# 12.4.13 Pipe

Consumption of pipe will be approximately the total length of drift and subdrift (2000 feet) and of raise (2400 feet):

4400 feet of 4 inch pipe at \$3.71 - \$ 16,300 4400 feet of 2 inch pipe at \$2.26 - 10,000

Fittings (25 percent of above) - 6,600

TOTAL ANNUAL COST - \$ 32,900

=====

#### 12.4.14 Small Tools

Small tools and miners' gear will cost from \$3.50 to \$4.00 per man-shift.

Annual cost (\$3.70 x 31 men x 250 days)	-	\$ 28,900
For the mechanical crew, including shop tool		
replacement (allowance)	-	10,000
TOTAL ANNUAL COST	-	\$ 38,900

#### 12.4.15 General Consumables

General consumables including engineering supplies, stationary, electrical replacements, lubricants, ventilation, and other costs, are estimated to be \$60,000 per year.

## 12.4.16 <u>Mine Maintenance Materials</u>

<u>Item</u>	Annual Cost
Rock Drills	\$ 41,300
Diesel Mobile Equipment	83,600
Electric Generators	16,400
Air Compressors	13,800
Pumps	2,500
Ventilation and Mine Air Heating	6,500
Slushers	22,900
Buildings	5,000
Small Vehicles	13,100
Electrical Miscellaneous	10,000
Mechanical Miscellaneous	10,000
TOTAL ANNUAL COST	\$ 225,100 ======

Total Monthly Cost = \$ 18,800 Cost per Ton = \$ 2.24

```
(a) Rock Drills
     Cost per foor drilled for drill maintenance,
     hose, and lubricants
                                                                     0.13
     Total footage drilled
                                                                   318,000 Feet
     TOTAL ANNUAL COST ($0.13 \times 318,000)
                                                                    41,300
(b) Diesel Mobile Equipment
     J.D.T. 413 13-Ton Trucks:
           Hourly Cost ($130,000 \times 0.7) ÷ 10,000
                                                                 $ 9.10
           Annual Cost ($9.10 \times 10 hours \times 250 days)
                                                                    22,700
     Two 2 cubic yard LHD units:
           Hourly Cost ($106,000 \times 0.8) \div 10,000
                                                                     8.48
          Annual Cost:
                (\$8.48 \times 2 \text{ units } \times 12 \text{ hours } \times 250 \text{ days}) - \$ 50,900
     Miscellaneous mobile equipment (mine service
     vehicle, surface equipment) allowance
                                                                 $ 10,000
     TOTAL ANNUAL COST
                                                                   83,600
(c) Electric Generators
     Allow $45.00 per day for generator maintenace:
     TOTAL ANNUAL COST ($45.00 x 365 days)
                                                                 $ 16,400
(d) Air Compressors
     Allow $55.00 per day for compressor maintenance:
     TOTAL ANNUAL COST ($55.00 x 250 days)
                                                                 $ 13,800
```

### (e) Pumps

Allow \$2500 per year for the cost of maintaining surface and underground pumps.

### (f) <u>Ventilation</u> and <u>Mine Air Heating</u>

Annual cost of fan maintenance	-	\$ 5,000
Annual cost of mine air heater maintenance	-	1,500
TOTAL ANNUAL COST	-	\$ 6,500

### (g) Slushers

Slusher and scraper maintenance cost is normally \$0.023 per cubic foot handled.

Cubic feet of raise muck (2400 ft x 5 ft x 8 ft) Cubic feet of stope muck (90,000 t x 10 $ft^3/t$ )		
TOTAL	-	\$ 996,000 ft <sup>3</sup>
Maintenance cost (\$0.023 v 996.000)	_	\$ 22 900

### (h) Buildings

Allow \$5000 per year for building maintenance and repair.

# (i) Small Vehicles

Allow 150 miles per day for a service truck and a flat deck truck. The operating cost will be \$0.35 per mile.

TOTAL ANNUAL COST (150 miles x 250 days x \$0.35) - \$ 13,100

# (j) Electrical Miscellaneous

Allow \$10,000 per year.

## (k) Mechanical Miscellaneous

Allow \$10,000 per year.

# 12.5 MILLING COSTS

## 12.5.1 General

The following milling costs include all the operations and services required at the mill site:

Item	Monthly Cost	Cost per Ton
Supervision and Technical Salaries	\$ 18,900	\$ 2.25
Operating Labour	53,200	6.33
Maintenance Labour	18,400	2.19
Operating Supplies	17,000	2.03
Maintenance Materials	28,100	3.35
Power	11,800	1.40
TOTAL MONTHLY COST	\$ 147,400 =======	-
Total Cost per Ton		\$ 17.55 =====

### 12.5.2 Supervision and Technical Salaries

<u>Position</u>	Number	Salary	Annual Cost
Mill Superintendent	1	\$ 50,000	\$ 50,000
General Foreman	1	38,000	38,000
Maintenance Foreman	1	35,000	35,000
Chief Assayer	1	32,000	32,000
Clerk	1	24,000	24,000
TOTAL Payroll Burden at 26.5 Percent	5		\$ 179,000 47,400
TOTAL ANNUAL COST			\$ 226,400

Total Monthly Cost = \$ 18,900 Cost per Ton Milled = \$ 2.25

## 12.5.3 Mill Operating Labour

Position	Number	Hourly Rate	Annual Salary	Annual Cost
Crusher Operator	1	\$ 12.50	\$ 26,000	\$ 26,000
Crusher Helper	1	11.00	22,880	22,880
Grinding/Flotation Operator	4	13.50	28,080	112,320
Filter Operator	4	12.50	26,000	104,000
Helper	4	11.50	23,920	95,680
Day Gang	4	11.00	22,880	91,520
Assayer	2	12.50	26,000	52,000
TOTAL	20			\$ 504,400
Payroll Burden at 26.5 Percent	t			133,700
TOTAL ANNUAL COST				\$ 638,100

Total Monthly Cost = \$ 53,200 Cost per Ton Milled = \$ 6.33

## 12.5.4 Maintenance Labour

Position	Number	Hourly Rate	Annual Salary	Annual Cost
Journeyman	<u>6</u>	\$ 14.00	\$ 29,120	\$ 174,720
TOTAL Payroll Burden at 26.5 Percen	6 t			\$ 174,720 46,280
TOTAL ANNUAL COST				\$ 221,000

Total Monthly Cost = \$ 18,400 Cost per Ton Milled = \$ 2.19

# 12.5.5 Mill Operating Supplies

### (a) Summary - Annual Cost

<u>Item</u>	Annual Cost
Reagents	\$ 111,700
Grinding Steel	58,200
Propane	9,600
Diesel Fuel	3,900
General Consumables	20,200
TOTAL ANNUAL COST	\$ 203,600 ======

Total Monthly Cost = \$ 17,000 Cost per Ton Milled = \$ 2.03

# (b) Reagents (Milling Rate: 100,800 Tons per Year)

<u>Item</u>	Addition or Usage Rate (1b/ton)	Annual Consumption (1b)	Cost (¢/1b)		ual st
Soda Ash	0.60	60,500	\$ 0.16	\$ 9	,600
Sodium Silicate	0.30	30,200	0.16	4	,800
Copper Sulfate	0.20	20,200	0.43	8	,600
Sodium Isopropyl Xanthate	0.15	15,100	0.82	12	2,300
Oleic Acid	1.50	151,200	0.40	60	,500
Quebracho	0.05	5,000	0.25	1	,300
Pine Oil	0.13	13,100	0.74	9	,700
TOTAL		295,300		\$ 106	,800
Total Transportatio	n Cost			4	,900
TOTAL ANNUAL REAGEN	T COST			\$ 111	,700

## (c) Grinding Steel

Item	Usage Ratio (1b/ton)	Annual Consumption (lb/year)	Cost (¢/1b)	Annual Cost
Grinding Balls (3")	1.00	100,000	0.28	\$ 28,000
Grinding Balls (1")	0.20	20,000	0.30	6,000
Mill Liners	0.15	15,000	1.05	15,750
Miscellaneous	-	<u>.</u>	-	5,000
TOTAL		135,000		\$ 54,750
Total Transportation	n Cost			3,450
TOTAL ANNUAL GRINDIN	IG STEEL C	OSTS		\$ 58,200

# (d) Mill Heating

It is estimated that 8400 gallons of propane will be required.

TOTAL ANNUAL COST (8400 x 1.14)

\$ 9,600

# (e) <u>Diesel Fuel</u>

Grader (400 hours at 2.1 gallons per hour)	-	840 gal
Front-End loader:		
(800 hours at 1.8 gallons per hour)	-	1,440 gal
TOTAL .	-	2,280 gal
TOTAL ANNUAL COST (\$1.70 x 2280)	-	\$ 3,900

### (f) General Consumables

General consumables such as tires, small vehicle operation, tools, stationary, etc., are estimated to cost \$0.20 per ton milled.

TOTAL ANNUAL COST 100,800 x 0.20

\$ 20,200

#### 12.5.6 Mill Maintenance Supplies

Mill maintenance supplies normally cost four percent of the mill equipment replacement cost.

Annual Cost (\$7,500,000 x 4 percent) \$ 300,000 Maintenance of surface equipment: Grader (400 hours at \$10.00) 4,000 Front-End loader (800 hours at \$10.00) 8,000 Miscellaneous mechanical maintenance 10,000 Miscellaneous electrical maintenance 10,000 Building maintenance 5,000 TOTAL ANNUAL COST \$ 337,000 Total Monthly Cost = \$ 28,100

Total Monthly Cost = \$ 28,100 Cost per Ton = \$ 3.35

#### 12.5.7 <u>Power</u>

Connected load in mill = 1284 horsepower (958 kilowatts)

Annual Consumption is:

958 kW x 90% Demand x 24 hours x 355 days x 0.8 Load Factor x 1.20 (20% allowance for other equipment) = 7,052,000 kilowatt hours.

TOTAL ANNUAL COST =  $7,052,000 \times \$0.02$  per kWH = \$141,000

Total Monthly Cost = \$ 11,800

Cost Per Ton = \$ 1.40