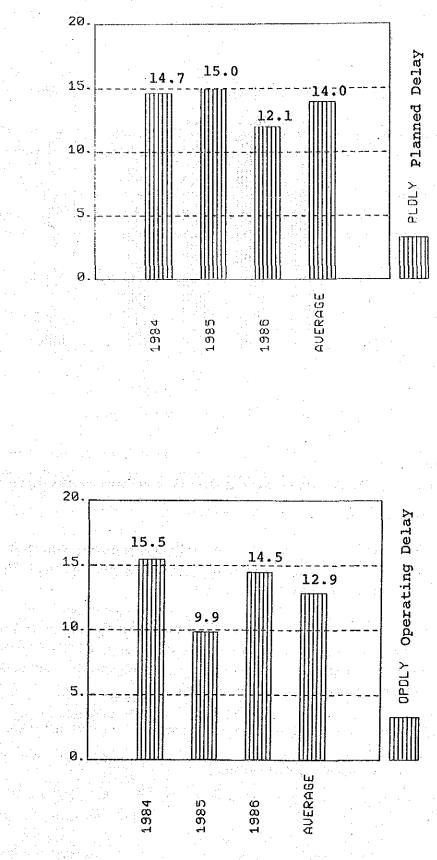
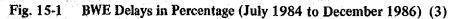


Fig. 15-1 BWE Delays in Percentage (July 1984 to December 1986) (2)





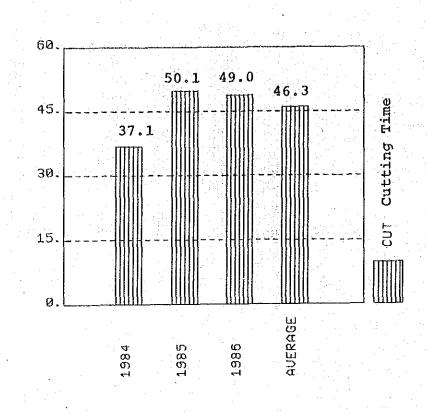


Fig. 15-1 BWE Delays in Percentage (July 1984 to December 1986) (4)

The operational delay is one of the scales indicating how efficiently mining systems are operated. It is 15.5, 9.9, 14.5% for 1984, 1985 and 1986 respectively maintaining almost constant level.

Cutting time is the time actually BWE's are cutting either waste or coal. It is 37.1% in 1984 which is considerably lower than 50.1 and 49.0% in 1985 and 1986 respectively. It may be interpreted that operators were still in the stage of learning in the first year of the operation.

The average for those three years is 46.3 and 49.6% for 1985 and 1986.

It has been aimed to achieve at 60%.

15-4 Productivity of "Run-of-Mine" Excavation

Main factors indicating productivity of the bucket wheel excavator system are the hourly cutting volume (BCM per hour) and actual cutting time of the system.

The "run-of-mine" extraction was extensively performed in 1984, however, the performance data is summarized in Table 15-4 from July to December excluding the trial operation period from the equipment start up in February and March to the end of June.

Table 15-4 shows the performance of the system.

As indicated in Table 15-4, the average productivity of the BWE system is as follows:

Waste	(BCM/hr)	635	
Coal	(BCM/hr)	575	i
	(Tons/hr)	748	1
Average	(BCM/hr)	629	J

The coal cutting rate is lower than that of waste cutting even though a whole seam extraction, so-called "run-of-mine" without segregations of mudstone plies.

The cutting time is 37.1% of the total time in 1984 from July to December.

Total yearly production scale of run-of-mine extraction in 1984 is estimated based on the actual conditions experienced from July to December in the same year.

24
37.1 (Actual 1984)
635
748
4
1:7.8

The annual production varies upon the yearly working days as shown in Fig. 15-2.

As indicated in Fig. 15-2, the estimated annual production is as indicated below, at the designated working days of 233 by SCC.

Waste Coal 4,752,000 BCM 609,300 tons

ne	1. 1. 191			۰ - ۲۰ - ۲۰ - ۲۰ - مستقدم - ۱ ۰			
	Jul.	Aug.	Sep.	Oct.	Nov.	Dec.	Total
Operating days Av. operating hrs./day Total hours	28.5 24.0 2,736.2	30.7 24.0 2.951.7					172.5 16,559.8
Maintenance delays Mechanical Electrical Sub-Total	589.1 69.4 658.5	563.1 53.5 616.6	272.6 127.6 400.2	92.7	150.7 68.5 219.2	245.0 60.0 305.0	2,088.1 471.7 2,559.8
CCP/Power failure	318.1	809,1 189,4	356.2	311.0 511.1			2,862.3 2,440.5
Operational delays	547.2	511.4		550.7	348.1	192.7	2,559.1
Cutting time (hrs.) Waste (incl. W.C.) Coal	1	110.9	120.5	122.1	71.9	35.5	5,536.2 601.9 6,138.1
	Operating days Av. operating hrs./day Total hours Maintenance delays Mechanical Electrical Sub-Total CCP/Power failure Planned delays Operational delays Cutting time (hrs.) Waste (incl. W.C.)	Jul. Operating days 28.5 Av. operating hrs./day 24.0 Total hours 2,736.2 Maintenance delays Mechanical 589.1 Electrical 69.4 Sub-Total 658.5 CCP/Power failure 318.1 Planned delays 257.1 Operational delays 547.2 Cutting time (hrs.) 0.0 Waste (incl. W.C.) 814.3 Coal 141.0	Jul. Aug. Operating days 28.5 30.7 Av. operating hrs./day 24.0 24.0 Total hours 2,736.2 2.951.7 Maintenance delays 889.1 563.1 Mechanical 589.1 563.1 Electrical 69.4 53.5 Sub-Total 658.5 616.6 CCP/Power failure 318.1 809.1 Planned delays 257.1 189.4 Operational delays 547.2 511.4 Cutting time (hrs.) 0.0 814.3 714.3 Coal 141.0 110.9 10.9	Jul.Aug.Sep.Operating days28.530.725.9Av. operating hrs./day24.024.024.0Total hours2,736.22.951.72,484.2Maintenance delays589.1563.1272.6Belectrical69.453.5127.6Sub-Total658.5616.6400.2CCP/Power failure318.1809.1356.2Planned delays257.1189.4150.7Operational delays547.2511.4409.0Cutting time (hrs.)0.00.0141.0Waste (incl. W.C.)814.3714.31,047.6Coal141.0110.9120.5	Jul.Aug.Sep.Oct.Operating days28.530.725.928.1Av. operating hrs./day24.024.024.024.0Total hours2,736.22.951.72,484.22.699.7Maintenance delays589.1563.1272.6267.6Blectrical69.453.5127.692.7Sub-Total658.5616.6400.2360.3CCP/Power failure318.1809.1356.2311.0Planned delays257.1189.4150.7511.1Operational delays547.2511.4409.0550.7Cutting time (hrs.)0.00.0141.0110.9120.5Qal141.0110.9120.5122.1	Jul.Aug.Sep.Oct.Nov.Operating days Av. operating hrs./day28.530.725.928.129.3Av. operating hrs./day Total hours24.024.024.024.024.02,736.22,951.72,484.22.699.72,808.0Maintenance delays Mechanical589.1563.1272.6267.6150.7Electrical69.453.5127.692.768.5Sub-Total658.5616.6400.2360.3219.2CCP/Power failure318.1809.1356.2311.0638.2Planned delays257.1189.4150.7511.1473.0Operational delays547.2511.4409.0550.7348.1Cutting time (hrs.) Waste (incl. W.C.)0.0814.3714.31,047.6844.51,057.8Coal141.0110.9120.5122.171.9	Jul.Aug.Sep.Oct.Nov.Dec.Operating days Av. operating hrs./day28.530.725.928.129.330.0Av. operating hrs./day Total hours24.024.024.024.024.024.02,736.22.951.72,484.22.699.72,808.02,880.0Maintenance delays Mechanical589.1563.1272.6267.6150.7245.0Electrical69.453.5127.692.768.560.0Sub-Total658.5616.6400.2360.3219.2305.0CCP/Power failure318.1809.1356.2311.0638.2429.7Planned delays257.1189.4150.7511.1473.0859.2Operational delays547.2511.4409.0550.7348.1192.7Cutting time (hrs.) Waste (incl. W.C.)0.0141.0110.9120.5122.171.935.5Coal141.0110.9120.5122.171.935.5

Table 15-4 BWE System Perforamnce in 1984 (July to December)

Production

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. . .

1

	Overburden total	a transform	1902.5	ata ay in				
Waste BCM (less waste co Waste coal	(less waste coal)	552,540	385,296	618,003	494,876	679,119	787,686	3,517,520
	Waste coal	0	285 205	0	0 404 876	0 679 119	787 686	0 3,517,520
	Total Waste	552,540	565,290	018,005	+/+,0/0	012,112	101,000	0,01.,020
•	Product coal	126,381	65,879	82,373	89,077	61,328	24,921	449,959
Coal, tons	Washable coal	0	0	0	0	0	0	0
, ,	Sub-Total	126,381	65,879	82,373	89,077	61,328	24,921	449,959
	Waste coal (waste)	0	0	0	0	0	0	0
	Product coal	97,216	50,676	63,364	68,521	47,175	19,170	346,122
BCM	Washable coal	0	0	0	0	0	0	0
	Sub-total	97,216	50,676	63,364	68,521	47,175	19,170	346,122
Total BCM	1 moved	649,756	435,972	681,367	563,397	726,294	806,856	3,863,642
Stripping	ratio	4.4	5.8	7.5	5.6	11.1	31.6	7.8

Cutting Rate

Cutting Rate		n na h Stair Stair S					
Waste (O.B.+W.C.) (BCM/hr) Coal(BCM/ht.) (tons/hr.)	679 689 896	539 457 594	590 526 684		642 656 853	745 540 702	635 575 748
Average (BCM/hr.)	680	528	583	583	643	738	629

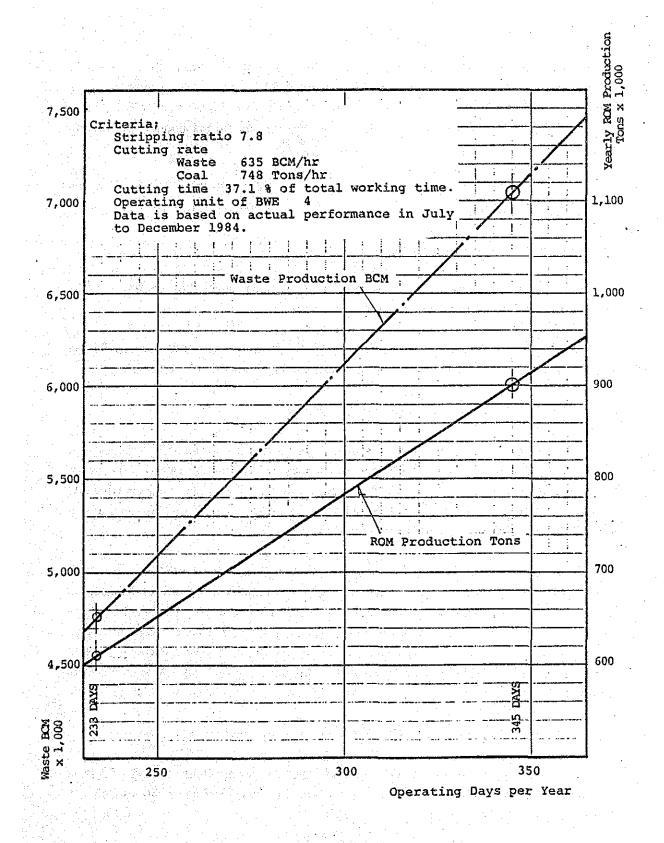


Fig. 15-2 Estimated Yearly Mine Production (ROM) in 1984

It must be noted that the actual operating time for 6 months from July to December 1984 is approximately 16,600 hours for the total of 4 BWE units. It means total working days is considered 172.5 days per 6 months, therefore, it may indicate that SCC operated the mine on a 345 days per year basis in 1984. In this case, the estimated yearly production turns out to be:

Waste	an a	7,037,000 BCM
Coal	te je	902,100 tons

15-5 Productivity of Selective Mining

The selective mining has been exclusively performed since 1985 to improve the quality of coal delivered to NAPOCOR Calaca power plant.

As previously mentioned in the section 14-4, cutting rate of BWE is considerably affected by cumbersome coal winning operations involving mudstone ply removals.

Table 15-5 shows the performance of the system.

The coal cutting rate of the BWE system is 612 tons per hour and 587 tons per hour in 1985 and 1986 respectively. Those cutting rates are approximately 140 to 160 tons per hour lower than that of 1984 actual 748 tons per hour. Comparing the average cutting rate of 599 tons per hour in 1985 and 1986 with the 748 tons per hour of 1984 actual, it is lowered by 150 tons per hour due to selective mining.

Since 1986, an emergence of hard sandstone which is almost unable to excavate with BWE has been experienced. The volume is roughly estimated 20% of the total waste. In the excavation of the hard sandstone, ripping with bulldozers and/or blasting have been employed prior to the BWE excavation. As a result, belting cut has been experienced from time to time. If the hard material increases in the future, it may be difficult to depend upon the system with belt conveyors, so that an introduction of truck and shovel system would be required.

In 1986, cutting rate is 601 BCM per hour in the hard material excavation.

The cutting rates of soft material shift with years showing slight increase, 635 BCM per hour, 698 BCM per hour and 774 BCM per hour in 1984, 1985 and 1986 respectively. The average is 720 BCM per hour.

Combining both hard and soft materials, the waste material cutting rates are 635

۰.		1.00	1.1		1. 1990 - 1990 - 1990 - 1990 - 1990 - 1990 - 1990 - 1990 - 1990 - 1990 - 1990 - 1990 - 1990 - 1990 - 1990 - 19	مر ومن مناسب	<u></u>	
	1985–1986	16,643	4,131 20,774	1,948	11,984,279 2,484,162 14,468,441	1,166,535	720 601 696	599
	19841986	22,179	4,131 26,310	2,548	15,501,807 2,484,162 17,985,969	1,616,494	699 601 684	634

Table 15-5 BWE Cutting Rate Comparison

2,484,162 4,847.5 4,131.0 8,978.5 579,278 3,751,215 774 601 694 587 6,235,377 987.1 1986 11,795.6 8,233,064 8,233,064 698 Ţ 959.3 698 612 11,795.6 587,257 1985 5,536.2 5,536.2 I. 635 601.9 3,517,528 635 748 3,517,528 449,959 1984 (7-12) Hard Waste (BCM/hr.) Total Waste (BCM/hr.) Soft Waste (BCM/hr.) Hard Waste (BCM) Soft Waste (BCM) Coal (tons/hr.) Material Sub-Total Sub-Total Coal (tons) Hard Waste Soft Waste Coal Material Moved Cutting Rate Cutting Time (hours)

Notes: 1984 (7–12) ROM 1985, 1986 Selective mining

BCM per hour, 698 BCM per hour and 694 BCM per hour in 1984, 1985 and 1986 respectively, showing slight improvement.

The average of the three years from 1984 to 1986 is 684 BCM per hour and 696 BCM per hour for the two years average from 1985 to 1986.

The total cutting rate of waste material is affected by the amount of hard material, so that the volume of the hard material must be quantified accurately to evaluate future mine production, by an implementation of detail geological study.

15-6 Production Estimate of BWE System

Annual production of the BWE system is estimated based on several assumptions derived from the performance in the past.

1) Run-of-Mine Extraction

Assuming that the cutting rates of BWE system are 720 BCM per hour, 601 BCM per hour and 748 tons per hour for soft waste material, hard waste material and coal respectively, the yearly production is estimated.

The cutting rates of the waste materials are based on the actual performance from 1985 to 1986. The coal cutting rate is the actual of run-of-mine extraction performed from June to December in 1984.

The estimate is done in the cases of stripping ratio 1 to 7.8 and 1 to 10.8. They are the actuals of the BWE operation in 1984 and 1986 respectively.

At each stripping ratio, four cases of BWE cutting time are assumed, which are as follows:

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(1) 37.1%	Actual in 1984, from July to December
(2) 46.3%	Actual from July 1984 to December 1986
(3) 49.6%	Actual from 1985 to 1986
(4) 54.8%	Same as (2) except CCP delay adopted 0.4%,
	considered to be the best attainable performance

Table 15-6 and 15-7 show the summary of the yearly run-of-mine production estimate in various cases.

Table 15-6 Yearly Production Estimate (1)

Run-of-Mine Extraction at Stripping Ratio 7.8

7.8 20.0 720	(Hard/Soft)	0.25	
720		0.25]]
	(Decod 295-296]
	(Daged 195 196)		
	(Daged '96' '97'		· · ·
		ictual selective	
601	mining data)	10010	
748	('84, 7-12 actua	I ROM)	J
			_
249			
(1)	(2)	(3)	(4)
1004 (7 10)	3947 396	105 105	'84.7'86
			Actual
Actual	Actuai	Actual	Except CCP
			DLY 0.4%
		· · ·	DL1 0.470
37.1	46.3	49.6	54.8
8,868	11,068	11,856	13,099
			11
			6,487,710
1 Contract of the second seco second second sec			1,621,927
			8,109,637
703,883	878,430	941,040	1,039,697
	(1) 1984 (712) Actual 37.1	(1) (2) 1984 (712) '84.7-'86 Actual Actual 37.1 46.3 8,868 11,068 4,392,227 5,481,404 1,098,057 1,370,351 5,490,284 6,851,756	(1) (2) (3) 1984 (7-12) '84.7-'86 '85-'86 Actual Actual Actual 37.1 46.3 49.6 3868 11,068 11,856 4,392,227 5,481,404 5,872,088 1,098,057 1,370,351 1,468,022 5,490,284 6,851,756 7,340,110

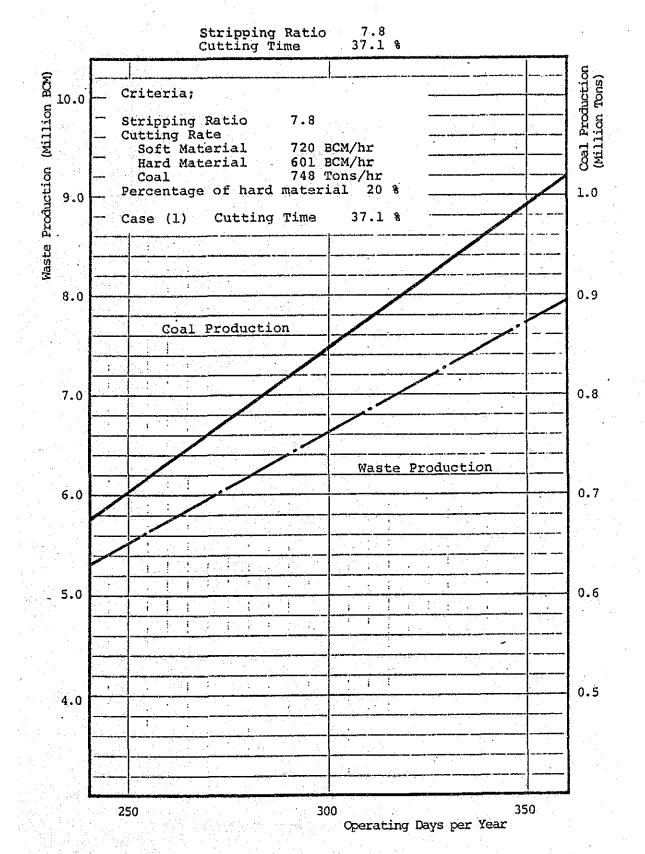
Operating Days/Year	301			
Cases	(1)	(2)	(3)	(4)
	1984 (7–12) Actual	'84.7–'86 Actual	'85'86 Actual	'84.7–'86 Actual Except CCP DLY 0.4%
Cutting Time (%) (Hours for 4 BWE)	37.1 10,720	46.3 13,379	49.6 14,332	54.8 15,835
Production Volume				
Soft Material (BCM/yr.)	5,309,479	6,625,115	7,098,387	7,842,573
Hard Material (BCM/yr.)	1,327,370	1,656,529	1,774,597	1,960,643
Sub-Total (BCM/yr.)	6,636,849	8,282,644	8,872,984	9,803,216
Coal (tons/yr.)	850,878	1,061,877	1,137,562	1,256,823

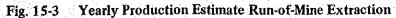
Operating Days/Year	360			
Cases	(1)	(2)	(3)	(4)
	1984 (7–12) Actual	'84.7–'86 Actual	'85'86 Actual	'84.7–'86 Actual Except CCP
		na an a		DLY 0.4%
Cutting Time (%) (Hours for 4 BWE)	37.1 12,822	46.3 16,001	49.6 17,142	54.8 18,939
D. Leater Mahama				
Production Volume Soft Material (BCM/yr.)	6,350,208	7,924,922	8,489,765 2,122,441	9,379,821 2,344,955
Hard Material (BCM/yr.) Sub-Total (BCM/yr.) Coal (tons/yr.)	1,587,552 7,937,759 1,017,661	9,906.153 1,270,020	10,612,207 1,360,539	2,344,933 11,724,777 1,503,177

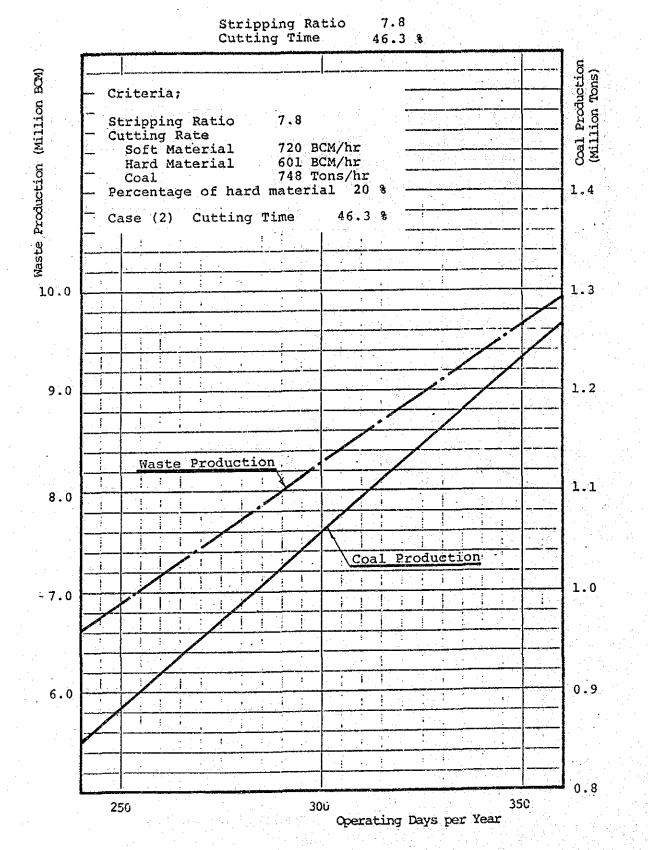
Table 15-6 Yearly Production Estimate (2)

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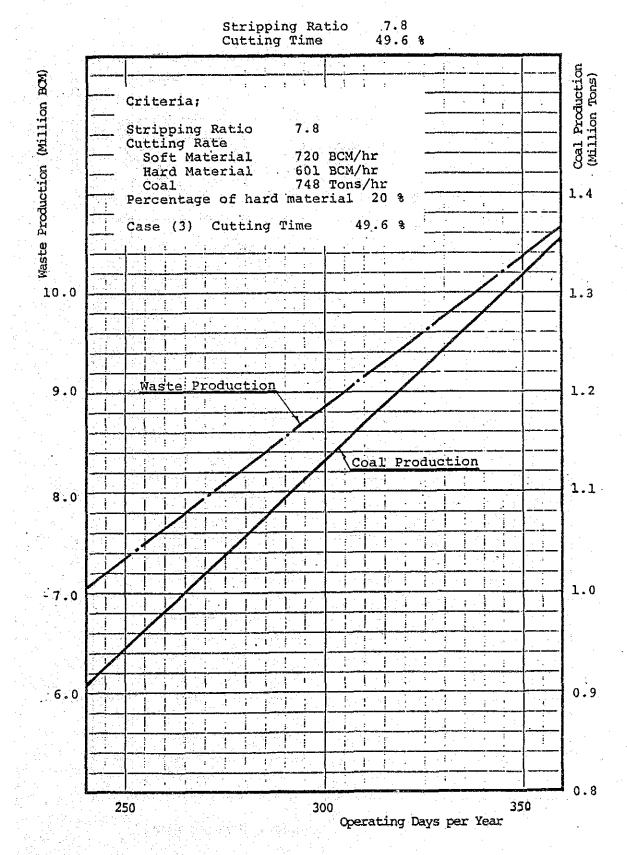
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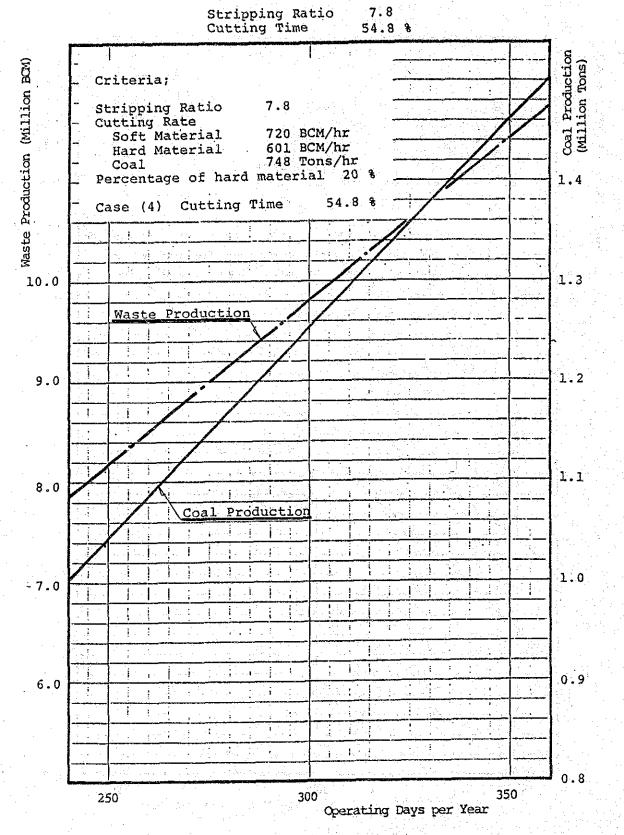


Fig. 15-6 Yearly Production Estimate Run-of-Mine Extraction

Table 15-7 Yearly Production Estimate (1)

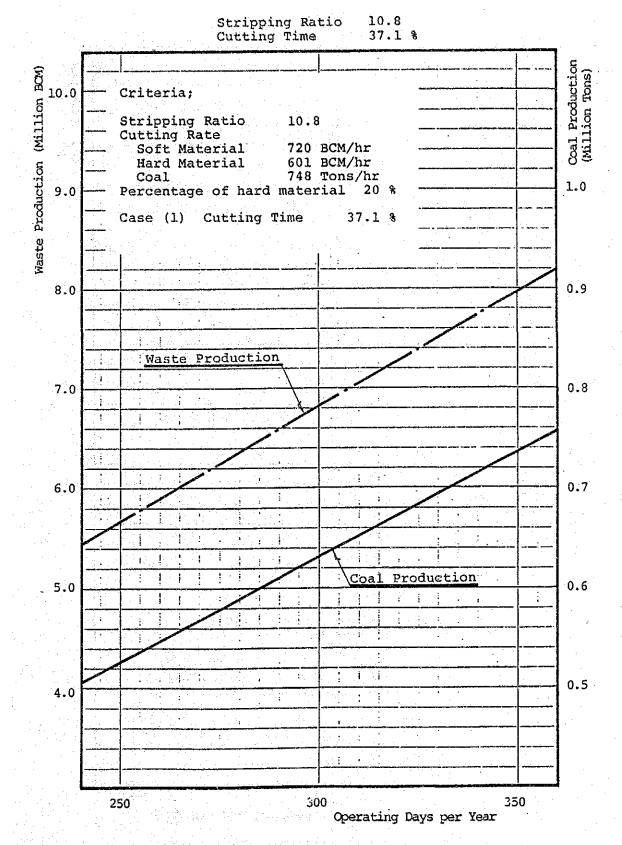
Run-of-Mine Extraction at Stripping Ratio 10.8

Stripping Ratio	10.8			
% of Hard Material	20.0	(Hard/Soft)	0.25	
Cutting Rate				
Soft Material (BCM/hr.)	720	(Based '85-'86	actual selective	
Hard Material (BCM/hr.)	601	mining data)		
Coal (tons/hr.)	748	('84.7-12 actua	1 ROM)	
			· · · · · · · · · · · · · · · · · · ·	•
Operating Days/Year	249			
	· · · · · · · · · · · · · · · · · · ·			
Cases	(1)	(2)	(3)	(4)
				10.1 8 10.4
	1984 (7–12)	'84.7–'86	'85–'86	'84.7 – '86
	Actual	Actual	Actual	Actual
	14 - A - A - A - A - A - A - A - A - A -			Except CCP
				DLY 0.4%
Cutting Time (%)	37.1	46.3	49.6	54.8
(Hours for 4 BWE)	8,868	11,068	11,856	13,099
Production Volume				
Soft Material (BCM/yr.)	4,525,519	5,637,875	6,050,423	6,684,742
Hard Material (BCM/yr.)	1,131,405	1,411,969	1,512,606	1,671,185
Sub-Total (BCM/yr.)	5,657,024	7,059,843	7,563,029	8,355,927
Coal (tons/yr.)	523,798	653,689	700,280	773,697

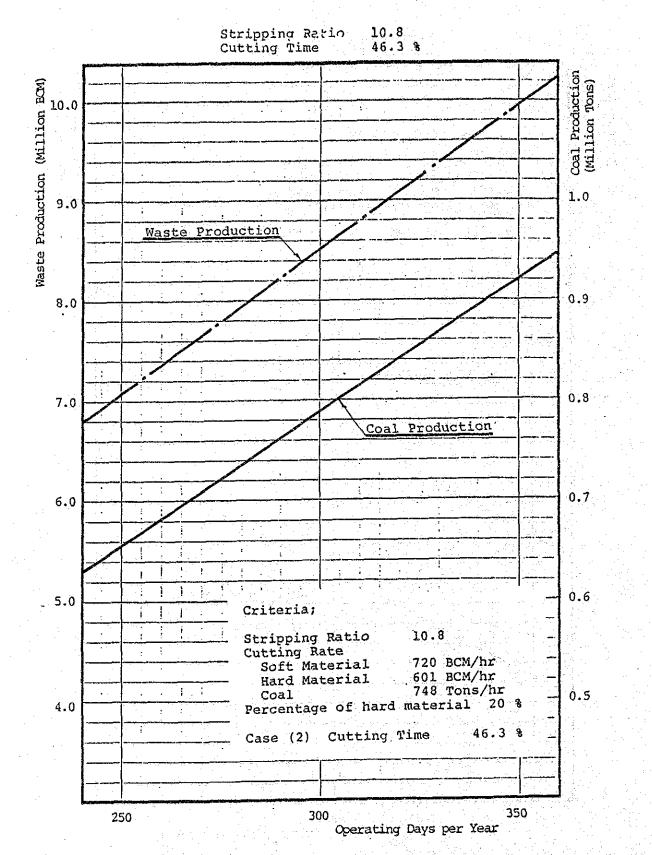
Operating Days/Year	301			
	· · · · · · · · · · · · · · · · · · ·			· · · · · · · · · · · · · · · · · · ·
Cases	(1)	(2)	(3)	(4)
	1984 (7–12) Actual	'84.7—'86 Actual	'85–'86 Actual	'84.7'86 Actual Except CCP DLY 0.4%
Cutting Time (%) (Hours for 4 BWE)	37.1 10,720	46.3 13,379	49.6 14,332	54.8 15,835
Production Volume Soft Material (BCM/yr.) Hard Material (BCM/yr.) Sub-Total (BCM/yr.) Coal (tons/yr.)	5,470,728 1,367,682 6,838,410 633,186	6,827,351 1,706,836 8,534,188 790,203	7,313,965 1,828,491 9,142,456 846,524	8,080,752 2,020,188 10,100,940 935,272

·			<u> </u>	
Operating Days/Year	360			
Cases	(1)	(2)	(3)	(4)
	1984 (7–12) Actual	'84.7–'86 Actual	*85*86 Actual	'84.7'86 Actual Except CCP DLY 0.4%
Cutting Time (%) (Hours for 4 BWE)	37.1 12,822	46.3 16,001	49.6 17,142	54.8 18,939
Production Volume Soft Material (BCM/yr.) Hard Material (BCM/yr.) Sub-Total (BCM/yr.)	6,547,063 1,635,766 8,178,829	8,165,602 2,041,400 10,207,002	8,747,600 2,186,900 10,934,499	9,664,687 2,416,172 12,080,858
Coal (tons/yr.)	757,299	945,093	1,012,454	1,118,598

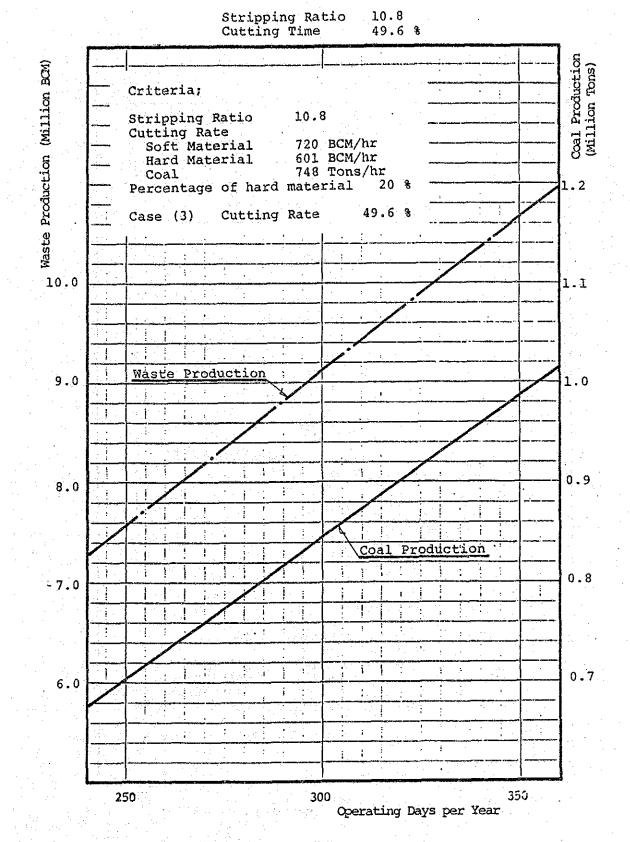
 Table 15-7
 Yearly Production Estimate (2)













Stripping Ratio Cutting Time 10.8 54.8 % Coal Production (Million Tons) Waste Production (Million BCM) ۰. . 1.2 Waste Production 10.0 1.1 ÷ 9.0 1.0 Coal Production 0.9 ,8.0 : : ; . . 1 0.8 - 7.0 Criteria; Stripping Ratio 10.8 Cutting Rate 720 BCM/hr Soft Material 0.7 601 BCM/hr Hard Material 6.0 748 Tons/hr Coal Percentage of hard material 20 % 54.8 % Case (4) Cutting Time 250 300 250 Operating Days per Year

Fig. 15-10 Yearly Production Estimate Run-of-Mine Extraction

Fig. 15-3 to 15-10 graphically indicate the estimated yearly run-of-mine production vs. yearly operating days at the above-mentioned two stripping ratios.

2) Selective Mining

3)

Yearly production of selective mining is estimated in the same manner as runof-mine extraction. The cutting rate of soft waste material and hard waste material is 720 BCM per hour and 601 BCM per hour respectively, which are same as run-of-mine extraction. The coal cutting rate of 599 tons per hour is adopted, which is the actual from 1985 to 1986 and is lower than 748 tons per hour for run-of-mine extraction.

The estimate is done on the stripping ratio 1 to 7.8 and 1 to 10.8, same as in run-of-mine extraction. At each stripping ratio, four cases of BWE cutting time are assumed same as in run-of-mine production estimate.

Table 15-8 and 15-9 are summary of the yearly selective mining production estimate in various cases.

Fig. 15-11 to 15-18 graphically indicate the estimated yearly selective mining production vs. yearly operating days at the above mentioned two stripping ratios.

Yearly Production Comparison (Run-of-Mine Extraction vs. Selective Mining)

The estimated yearly production are compared between run-of-mine extraction and selective mining on the case of stripping ratio of 1 to 10.8 as shown in Table 15-10 to 15-12.

Fig. 15-19 to 15-20 graphically indicate the results.

There is not much difference observed in the yearly production estimate between the two mining methods. As stripping ratio increases, the difference becomes much smaller, since the percentage of the time BWE's are engaged in coal cutting becomes much lower.

Table 15-8 Yearly Production Estimate (1)

Selective Mining at Stripping Ratio 7.8

.

Stripping Ratio	7.8	
% of Hard Material	20.0	(Hard/Soft) 0.25
Cutting Rate		
Soft Material (BCM/hr.)	720	(Based '85-'86 actual selective
Hard Material (BCM/hr.)	601	mining data)
Coal (tons/hr.)	599	
		and the stand of the second stand

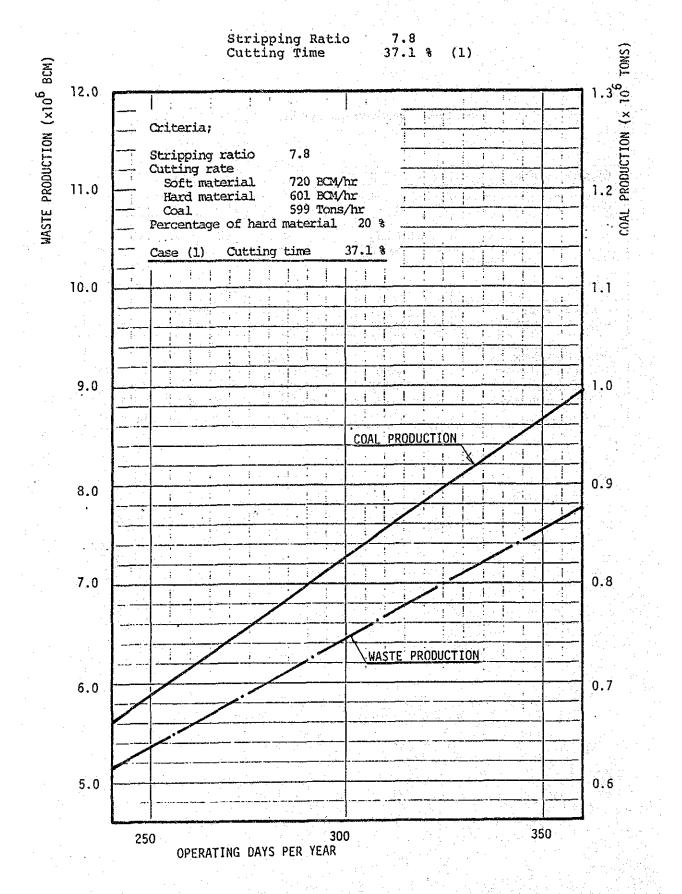
Operating Days/Year	249	(5 days/week)		
	(1)		(2)	(4)
Cases and a second second second	(1)	(2)	(3)	(4)
	1984 (7-12)	'84.7_'86	'85 –'86	' 84.7–'86
	Actual	Actual	Actual	Actual
		et of the second		Except CCP
			an a	DLY 0.4%
Cutting Time (%)	37.1	46.3	49.6	54.8
(Hours for 4 BWE)	8,868	11,068	11,856	13,099
Production Volume				
Soft Material (BCM/yr.)	4,279,277	5,340,446	5,721,083	6,320,873
Hard Material (BCM/yr.)	1,069,819	1,335,112	1,430,271	1,580,218
Sub-Total (BCM/yr.)	5,349,097	6,675,558	7,151,353	7,901,092
Coal (tons&yr.)	685,782	855,8421	916,840	1,012,960
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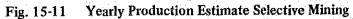
		analar <u>shina ara shina</u>		
Operating Days/Year	301	(6 days/week)	1997 - 1998 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - <u>1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997</u> - 1	
		an an an thug a the		
Cases	(1)	(2)	(3)	(4)
	1984 (7–12) Actual	'84.7—'86 Actual	'85–'86 Actual	'84.7–'86 Actual Except CCP DLY 0.4%
Cutting Time (%) (Hours for 4 BWE)	37.1 10,720	46.3 13,379	49.6 14,332	54.8 15,835
		an de la companya de		
Production Volume Soft Material (BCM/yr.) Hard Material (BCM/yr.) Sub-Total (BCm/yr.) Coal (tons/yr.)	5,172,942 1,293,235 6,446,177 828,997	6,455,720 1,613,930 8,069,650 1,034,570	6,915,847 1,728,962 8,644,808 1,108,309	7,640,895 1,910,224 9,551,119 1,224,502

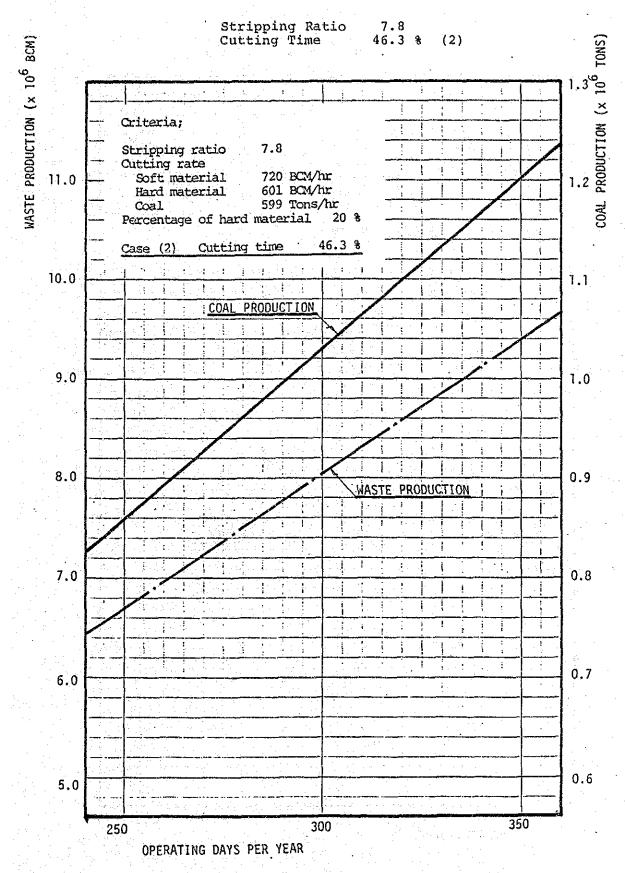
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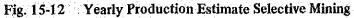
Table 15-8 Yearly Production Estimate (2)

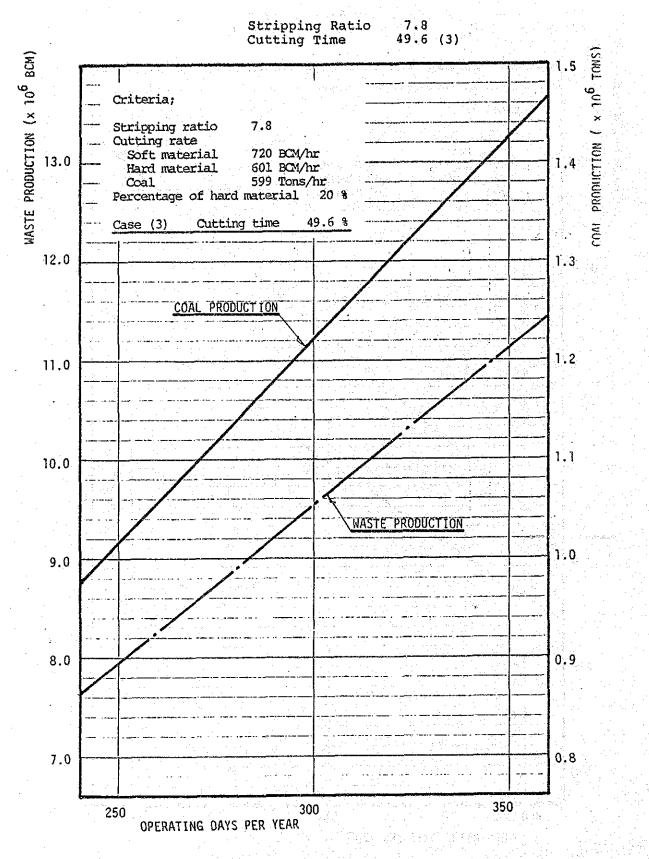
Operating Days/Year	360			
Cases	(1)	(2)	(3)	(4)
	1984 (7-12) Actual	'84.7–'86 Actual	'85–'86 Actual	'84.7—'86 Actual Except CCP DLY 0.4%
Cutting Time (%) (Hours for 4 BWE)	37.1 12,822	46.3 16,001	49.6 17,142	54.8 18,939
Production Volume Soft Material (BCM/yr.) Hard Material (BCM/yr.) Sub-Total (BCM/yr.) Coal (tons/yr.)	6,186,907 1,546,727 7,733,634 991,492	7,721,127 1,930,282 9,651,408 1,237,360	8,271,445 2,067,861 10,339,306 1,325,552	9,138,612 2,284,653 11,423,265 1,464,521

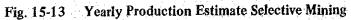












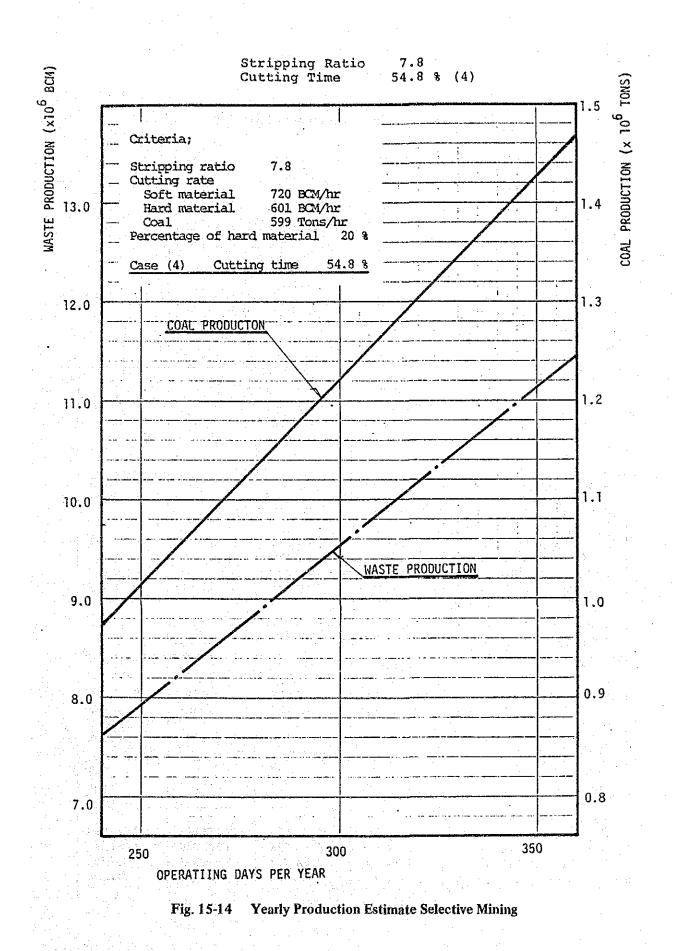




Table 15-9 Yearly Production Estimate (1)

 $\frac{1}{2}$

Selective Mining at Stripping Ratio 10.8

Stripping Ratio	10.8	
% of Hard Material	20.0	(Hard/Soft) 0.25
Cutting Rate		
Soft Material (BCM/hr.)	720	(Based '85-'86 actual selective
Hard Material (BCM/hr.)	601	mining data)
Coal (tons/hr.)	599	
	1	

Operating Days/Year 249

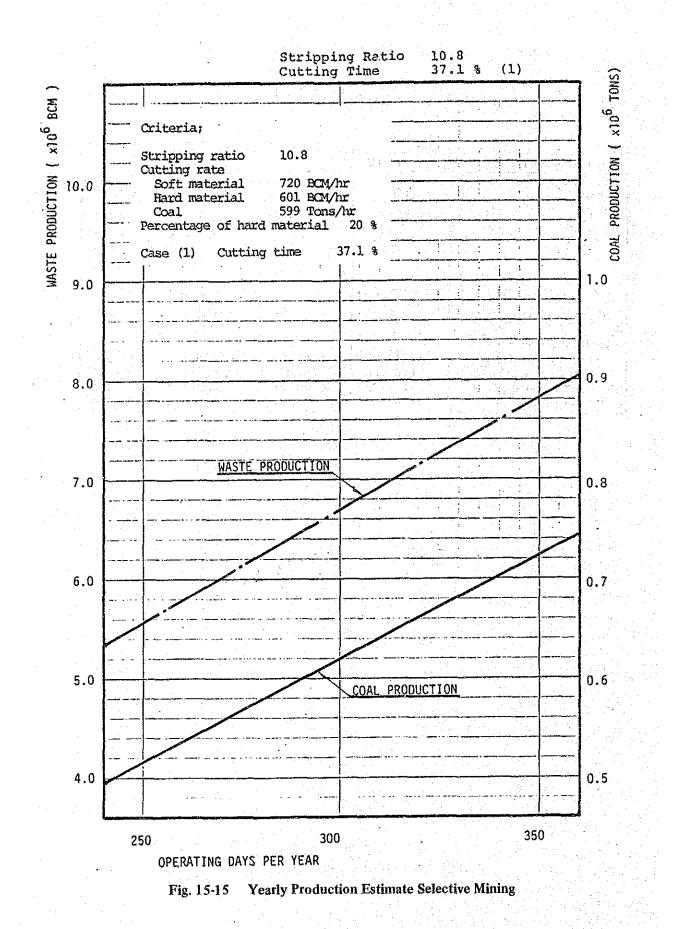
				e selação de tête
Cases	(1)	(2)	(3)	(4)
	1984 (7–12) Actual	'84.7–'86 Actual	'85—'86 Actual	'84.7–'86 Actual
				Except CCP DLY 0.4%
Cutting Time (%) (Hours for 4 BWE)	37.1 8,868	46.3 11,068	49.6 11,856	54.8 13,099
li Andrea Averantia		t		
Production Volume Soft Material (BCM/yr.)	4,438,441	5,539,078	5,933,872	6,555,972
Hard Material (BCM/yr.)	1,109,610	1,384,770	1,483,468	1,638,993
Sub-Total (BCM/yr.) Coal (tons/yr.)	5,548,051 513,708	6,923,848 641,097	7,417,340 686,791	8,194,964 758,793

Operating Days/Year	301	(6 days/week)		
Cases	(1)	(2)	(3)	(4)
	1984 (7–12) Actual	'84.7–'86 Actual	'85–'86 Actual	'84.7–'86 Actual Except CCP DLY 0.4%
Cutting Time (%) (Hours for 4 BWE)	37.1 10,720	46.3 13,379	49.6 14,332	54.8 15,835
Production Volume Soft Material (BCM/yr.) Hard Material (BCM/yr.) Sub-Total (BCM/yr.) Coal (tons/yr.)	5,365,344 1,341,336 6,706,680 620,989	6,695,833 1,673,958 8,369,792 774,981	7,173,074 1,793,269 8,966,343 830,217	7,925,090 1,981,273 9,906,363 917,256

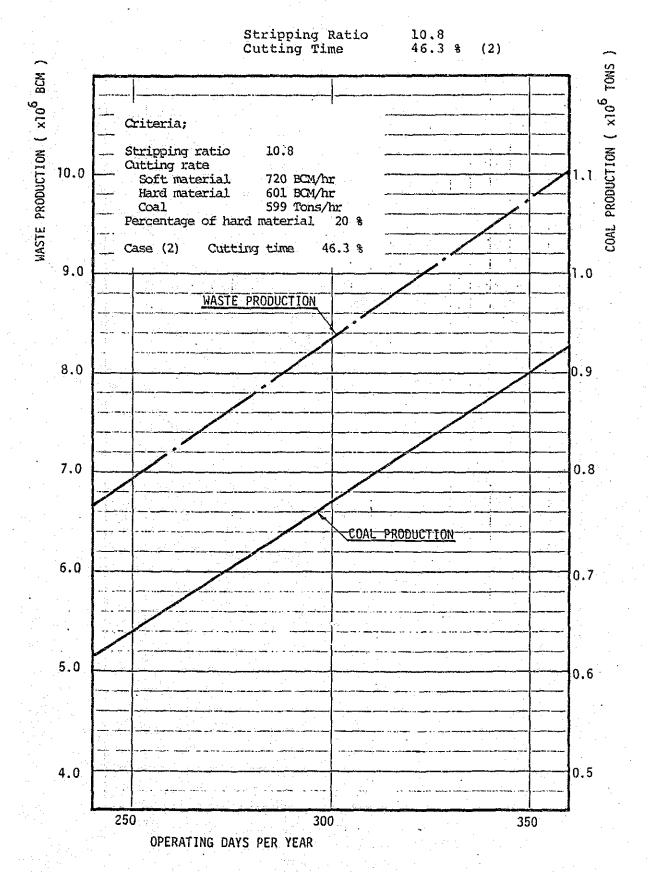
Table 15-9 Yearly Production Estimate (2)

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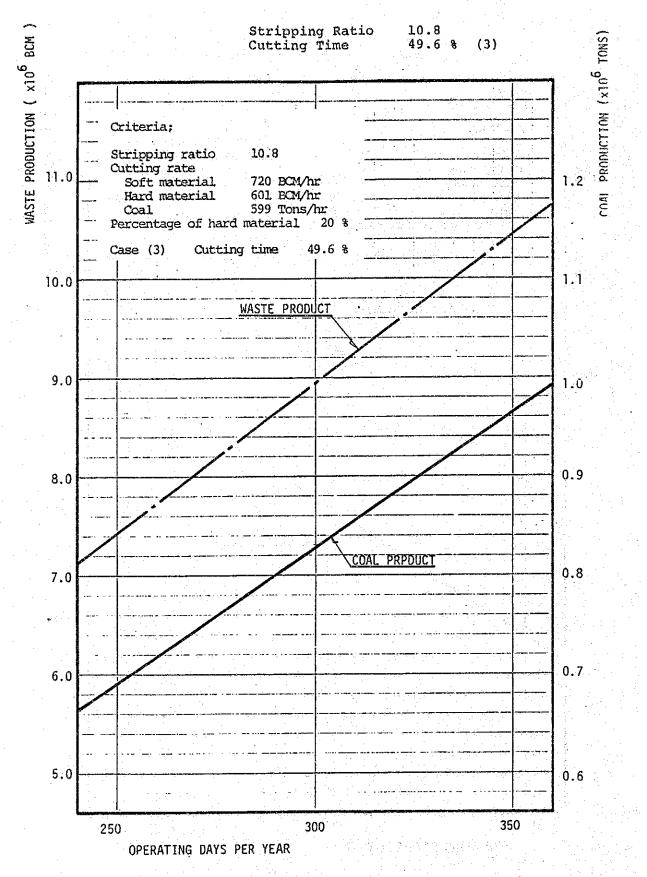
Operating Days/Year	360			
Cases	(1)	(2)	(3)	(4)
	1974 (7–12) Actual	'84.7–'86 Actual	'85'86 Actual	'84.7—'86 Actual Except CCP DLY 0.4%
Cutting Time (%) (Hours for 4 BWE)	37.1 12,822	46.3 16,001	49.6 17,142	54.8 18,939
Production Volume				
Soft Material (BCM/yr.) Hard Material (BCM/yr.) Sub-Total (BCM/yr.) Coal (tons/yr.)	6,417,023 1,604,256 8,021,278 742,711	8,008,306 2,002,076 10,010,382 926,887	8,579,092 2,144,773 10,723,865 992,950	9,478,512 2,369,628 11,848,141 1,097,050

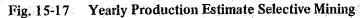


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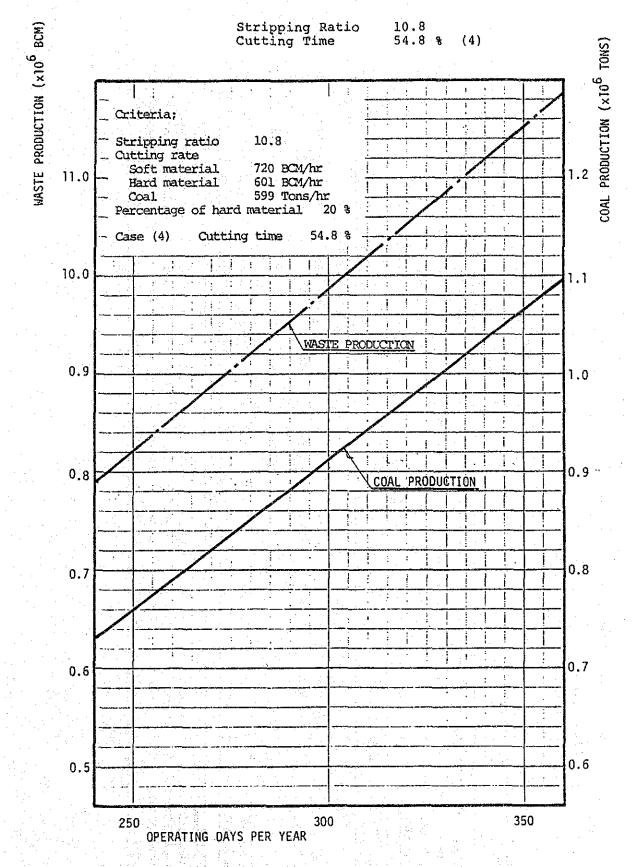


Fig. 15-18 Yearly Production Estimate Selective Mining

Table 15-10 Yearly Production Comparison (Selective Mining vs. Run-of-Mine Mining)

20% (1986 ACTUAL) 1:10.8 (1986 ACTUAL) (1) STRIPPING RATIO(2) HARD MATERIAL ASSUMPTIONS:

SELECTIVE MINING

Stripping Ratio	10.8		
% of Hard Mat'l	20.0	(Hard/Soft)	0.25
Cutting Rate			
Soft Material (BCM/Hr)	720	(Based '85-	(Based '85-'86 Actual

0.25

10.8 20.0 (Hard/Soft)

1 1 1

RUN-OF-MINE MINING

('84, 7-12 ROM Actual)

(Based '85-'86 Actual Selective Mining Data)

720 601 748

249

Culture nate			
Soft Material (BCM/Hr)	720	(Based '85-'86 Actual	••••••
Hard Material (BCM/Hr)	601	Selective Mining	
Coal (Tons/Hr)	599	Data)	
			1
Operating Days/Year	249		r
			ł

						•	•	
Cases	(1)	(2)	(3)	(4)	(1)	(2)	(3)	(4)
	1084 (7-17)	38, 7 - <u>3</u> 86	784 - 284	70. - 2 78,	1 024 /7 _171	7, 7, 84 7, 7, 86	78. <u></u> 58.	98, 2 95,
	Actual	Actual	Actual	Actual	Actual	Actual	Actual	Actual
				Except CCP				Except CCP
				Dly 0.4%				Dhy 0.4%
Cutting Time (%)	37.1	46.3	49.6	54.8 54.8	37.1	46.3	49.6	54.8
(Hours for 4 BWE)	8,868	11,068	11,856	13,099	8,868	11,068	11,856	13,099
							1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 -	
Production Volume Soft Material (BCM/Yr)	4,438,441	5,539,078	5,933,872	6,555,972	4,525,619	5,647,875	6,050,423	6,684,742
Hard Material (BCM/Yr)	1,109,610	1,384,770	1,483,468	1,638,993	1,131,405	1,411,969	1,512,606	1,671,185
								1

8,355,927 773,697

700,280 7,563,029

653,689

5,657,024

8,194,964 758,793

686,791

1,483,468 7,417,340

1,384,770 6,923,848 641,097

1,109,610 5,548,051

Sub-Total (BCM/Yr)

Coal (Tons/Yr)

513,708

523,798

1,411,969 7,059,843

Except CCP 54.8 15,835 8,080,752 2,020,188 84.7-786 10,100,940 935,272 Dly 0.4% Actual **£** 7,313,965 1,828,491 9,142,456 14,332 49.6 0.25 846,524 85-186 Actual ('84.7-12 ROM Actual) Selective Mining Data) (Based '85-'86 Actual ତି Yearly Production Comparison (Selective Mining vs. Run-of-Mine Mining) 10.8 20.0 | (Hard/Soft) 8,534,188 790,203 46.3 13,379 6,827,361 1,706,838 84.7-'86 Actual 3 RUN-OF-MINE MINING 1984 (7-12) 720 5,470,728 1,367,682 633,186 748 10,720 301 37.1 6,838,410 Actual Ξ Except CCP 54.8 1,981,273 15,835 197,256 84.7-786 9,906,363 Diy 0.4% 7,925,090 Actual () 0.25 7,173,074 1,793,269 49.6 14,332 8,966,343 830,217 85-786 Actual (Based '85-'86 Actual ම Selective Mining 1:10.8 (1986 ACTUAL) 20% (1986 ACTUAL) 10.8 20.0 [(Hard/Soft)] Data) 8,369,792 774,981 1,673,958 46.3 13,379 84.7-'86 6,695,833 Actual ଞ 1984 (7-12) 720 599 37.1 5,365,344 1,341,336 **Table 15-11** 301 10,720 620,989 6,706,680 Actual (1) STRIPPING RATIO (2) HARD MATERIAL Э Hard Material (BCM/Hr) Hard Material (BCM/Yr) Soft Material (BCM/Hr) Soft Material (BCM/Yr) Sub-Total (BCM/Yr) SELECTIVE MINING Operating Days/Year Production Volume Stripping Ratio % of Hard Mat7 (Hours for 4 VWE) Coal (Tons/Yr) ASSUMPTIONS: Cutting Time (%) Coal (Tons/Hr) Cutting Rate Cases

Table 15-12 Yearly Production Comparison (Selective Mining vs. Run-of-Mine Mining)

1:10.8 (1986 ACTUAL) 20% (1986 ACTUAL) ASSUMPTIONS: (1) STRIPPING RATIO (2) HARD MATERIAL

SELECTIVE MINING

	(Soft) 0.25	
	(Hard	
I	20.0	
Stripping Ratio	% of Hard Mat1	

0.25

----- <u>20.0</u> - <u>7</u> (Har6/Soft) - ⁷

RUN-OF-MINE MINING

(84.7-12 ROM Actual)

720 601 748

360

(Based '85-'86 Actual Selective Mining Data)

••••	
Soft Material (Bcm/Hr) 720	0 (Based '85-'86 Actual
Hard Material (Bcm/Hr) 601	l Selective Mining
Coal (Tons/Hr) 599	9 Data)

	(0)	(2)	(3)	(4)	(1)	(2)	(3)	(4)	
	1984 (7–12)	·84.7–'86	.85–.86	*84.7-*86	1984 (7–12)	. 84.7–36	* 85 *86	'84.7-'86	
	Actual	Actual	Actual	Actual	Actual	Actual	Actual	Actual	:
				Except CCP				Except CCP	
÷				Diy 0.4%				DIY 0.4%	
	37.1	46.3	49.6	54.8	37.1	46.3	49.6	54.8	
·	12,822	16,001	17,142	18,939	12,822	16,001	17,142	18,939	
т. 	6,417,023	8,008,306	8,579,092	9,478,513	6,543,063	8,165,602	8,747,600	9,664,687	:
	1,604,256	2,002,076	2,144,773	2,369,628	1,635,766	2,041,400	2,186,900	2,416,172	
	8,021,278	10,010,382	10,723,865	11,848,141	8,178,829	10,207,002	10,934,499	12,080,858	
		-							

1,118,598

1,012,454

945,093

757,299

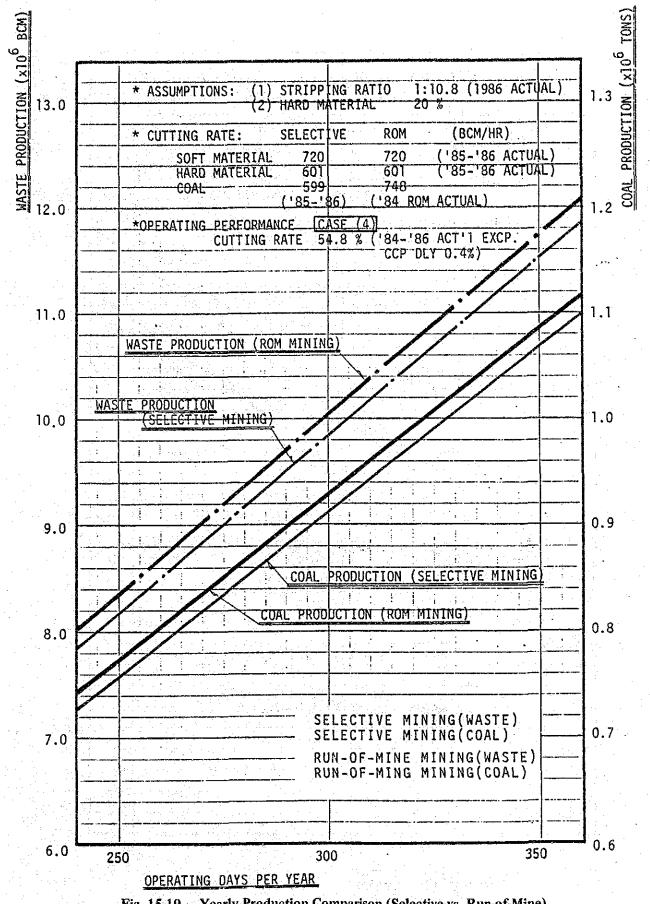
1,097,050

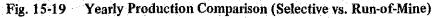
992,950

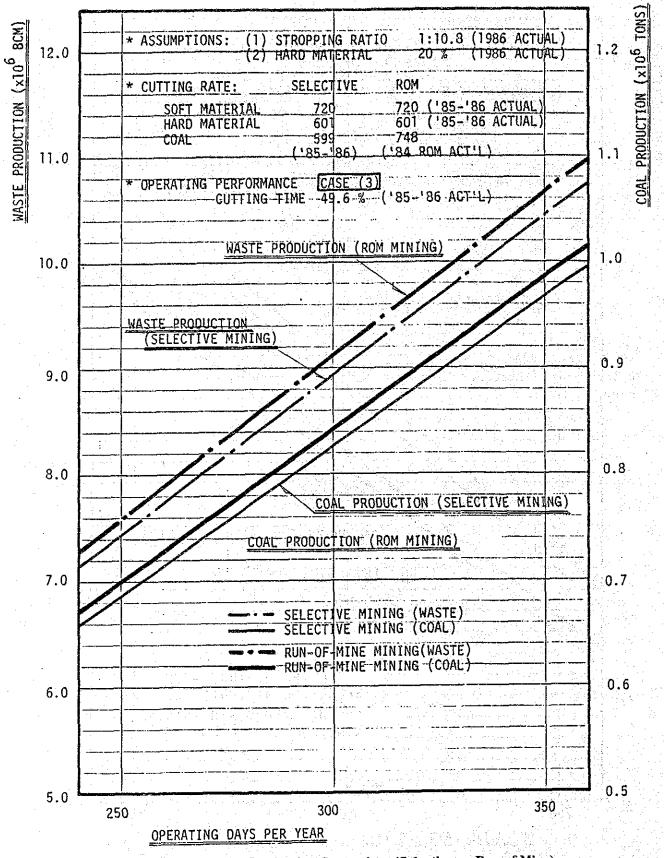
926,887

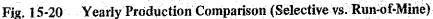
742,711

Coal (Tons/Yr.)









4) Summary

The results of both run-of-mine extraction and selective mining are summarized in Table 15-13.

The production scenario which is most likely occurs is the operation with 46.3% cutting time at the stripping ratio of 1 to 10.8. Currently only selective mining coal is accepted, so that the 641,000 tons of selective mining coal is likely produced by removing approximately 6,900,000 BCM of waste at stripping ratio of 1 to 10.8, with 249 days/year.

If run-of-mine extraction was implemented, the percentage of cutting time would be better than that of selective mining due to simpler coal winning procedures. As a result, the cutting time of 49.6% could be attained. In this case, 700,000 tons of coal production is expected with approximately 7,500,000 BCM of waste removal at stripping ratio of 1 to 10.8.

In any cases, a little amount of additonal coal production is expected from conventional mining system.

It is noted, however, that more detailed geological and mine planning studies must be implemented to verify the yearly stripping ratio, since the production scale to be achieved by the existing mining system is almost decided by the stripping ratio.

Operating Day	an an Angelan an Angelan Angelan	249 (5 days per w	veek operation)	
Stripping Ratio	7.8	and a state of the	10.8	97 } }
Mining System	ROM	Selective		Selective
Cutting Time	an a			
(1) 37.1%	W 5,490	W 5,349	W 5,657	W 5,548
(1) 5/11/5	C 704	• .		C 514
en el Colora de Latin de Alacha.				
(2) 46.3%	W 6,852		W 7,060	
	C 878	C 856	C 654	C 641
(3) 10 6%	W 7,340	W 7,151	W 7 563	W 7,417
(3) 49.6%	C 941	C 917	C 700	- 1 - 1 - 1 - E - 1 - 1 - 1 - 1 - 1 - 1
the second se		W 7.901	W 8,356	W 8,195
(4) 54.8%				
Operating Day	C 1,040	C 1,013	C 774	
	C 1,040	C 1,013	C 774	C 759
Operating Day	C 1,040	C 1,013 310 (6 days per w	C 774 veek operation)	C 759
Operating Day Stripping Ratio	C 1,040 7.8	C 1,013 310 (6 days per w	C 774 veek operation) 10.8	C 759
Operating Day Stripping Ratio Mining System Cutting Time	C 1,040 7.8 ROM	C 1,013 310 (6 days per w Selective	C 774 veek operation) 10.8 ROM	C 759 Selective
Operating Day Stripping Ratio Mining System	C 1,040 7.8 ROM W 6,637	C 1,013 310 (6 days per w Selective W 6,466	C 774 veek operation) 10.8 ROM W 6,838	C 759 Selective W 6,707
Operating Day Stripping Ratio Mining System Cutting Time	C 1,040 7.8 ROM	C 1,013 310 (6 days per w Selective	C 774 veek operation) 10.8 ROM	C 759 Selective W 6,707
Operating Day Stripping Ratio Mining System Cutting Time (1) 37.1%	C 1,040 7.8 ROM W 6,637	C 1,013 310 (6 days per w Selective W 6,466	C 774 veek operation) 10.8 ROM W 6,838	C 759 Selective W 6,707
Operating Day Stripping Ratio Mining System Cutting Time	C 1,040 7.8 ROM W 6,637 C 851	C 1,013 310 (6 days per w Selective W 6,466 C 829	C 774 veek operation) 10.8 ROM W 6,838 C 633	C 759 Selective W 6,707 C 621
Operating Day Stripping Ratio Mining System Cutting Time (1) 37.1%	C 1,040 7.8 ROM W 6,637 C 851 W 8,283 C 1,062	C 1,013 310 (6 days per w Selective W 6,466 C 829 W 8,070 C 1,035	C 774 veek operation) 10.8 ROM W 6,838 C 633 W 8,534 C 790	C 759 Selective W 6,707 C 621 W 8,370 C 775
Operating Day Stripping Ratio Mining System Cutting Time (1) 37.1%	C 1,040 7.8 ROM W 6,637 C 851 W 8,283 C 1,062 W 8,873	C 1,013 310 (6 days per w Selective W 6,466 C 829 W 8,070 C 1,035 W 8,645	C 774 veek operation) 10.8 ROM W 6,838 C 633 W 8,534 C 790 W 9,142	C 759 Selective W 6,707 C 621 W 8,370 C 775 W 8,966
Operating Day Stripping Ratio Mining System Cutting Time (1) 37.1% (2) 46.3%	C 1,040 7.8 ROM W 6,637 C 851 W 8,283 C 1,062	C 1,013 310 (6 days per w Selective W 6,466 C 829 W 8,070 C 1,035	C 774 veek operation) 10.8 ROM W 6,838 C 633 W 8,534 C 790	C 759 Selective W 6,707 C 621 W 8,370 C 775
Operating Day Stripping Ratio Mining System Cutting Time (1) 37.1% (2) 46.3%	C 1,040 7.8 ROM W 6,637 C 851 W 8,283 C 1,062 W 8,873	C 1,013 310 (6 days per w Selective W 6,466 C 829 W 8,070 C 1,035 W 8,645	C 774 veek operation) 10.8 ROM W 6,838 C 633 W 8,534 C 790 W 9,142	C 759 Selective W 6,707 C 621 W 8,370 C 775 W 8,966

Table 15-13Summary of Mine Production (by 4-BWE System)

Coal Reserves in Unong Pit

16.

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- 16-1 Original Reserves
- 16-2 Remaining Reserves (As of the end of 1986)

16-3 Summary

16. Coal Reserves in Unong Pit

16-1 Original Reserves

이 물건 수는 물을 두 같은 것이 가지?

Coal reserve estimation in the Unong pit is stated in the feasibility report for the Semirara Coal Project (Unong Mine) prepared by Austromineral Ges.m.b.H., Vienna, Austria, June 1980. Several geologic sections showing the occurrence of the coal seams are attached to the said feasibility report as back-up data.

In the feasibility study report, the coal reserves are calculated inside of the designed Unong pit area down to the depth of 150 m below sea level by using modified polygone method. In the reserve calcualtion, the whole pit area was divided into triangle coal blocks, each of which had more than three drill holes. The coal reserve tonnage in the respective coal block was obtained by a multiplication of the surface area, average coal seam thickness in the block and specific gravity of coal which was estimated 1.3. The total reserve in the Unong pit is an aggregation of reserves in the all coal blocks.

In the feasibility report, 17,220,000 tons of total reserves, comprising 13,945,000 tons of proven reserve and 3,275,000 tons of probable to proven reserve, is estimated. The 3,400,000 tons of dirty coal with low calorific value ranging from 1,800 to 3,500 Btu/lb is not included in the total reserve. The estimated specific gravity is 1.7 for the dirty coal.

Including the dirty coal, the whole reserves in the Unong pit turns out to be 20,620,000 tons. However, if the dirty coal is blended with saleable coal, it would be very difficult to meet the contracted coal specifications to NAPOCOR, so that it should not be included in the mineable reserve.

It is also mentioned that there is an additional coal field where approximately 731,000 tons of reserves were expected on the immediate south of the designed pit limit.

The coal reserves in the Unong pit, estimated by Austromineral, is summarized in the Table 16-1.

The proven reserves and the probable to proven reserves are represented by the Main seam and the Minor seams respectively, according to Austromineral.

11 July 18

The Main seam which occupies approximately 81% of the total reserves according to the feasibility report comprises 14 plies as indicated in Fig. 13-2. Amongst those plies, #2, #8 and #10 are mudstone, so-called clay, which thickness ranges from 1 to 3 meters. It is not clearly mentioned in the feasibility report whether the calculated reserve includes

Reserves	Tons (Ratio)
Proven (Main seam) Probable to Proven (Minor seams)	13,945,000 (81%) 3,275,000 (19%)
Total	17,220,000 (100%)
Dirty Coal in Minor Seam	3,400,000
Grand Total	20,620,000
Coal reserve in the additional field on the south side of the designed pit limit	731,000

Table 16-1 Unong Pit Coal Reserve (In-situ) by Austromineral

Unit: Metric Tons

Note: Specific gravity of coal Specific gravity of dirty coal

2)

those mudstone plies or not, but it seems to include all plies from the top to the bottom of the seam. It is so-called "run-of-mine".

1.7

Definitions of the proven and probable reserves are not mentioned, either, however, they are generally known as follows:

- 1) Proven reserves are those in which the reserves have been blocked out in three dimensions by excavation or drilling, but include in addition, minor extensions beyond actual openings and drill holes, where the geological factors that limit the reserve are definitely known and where the chance of failure of the reserve to reach those limit is so remote as not to be factor in the practical planning of mine operations.
 - Probable reserves cover extensions near at hand to proven reserves where the conditions are such that reserves will probably be found but where the extent and limiting conditions cannot be so precisely defined as for proven reserve. Probable coal reserves may also include reserve that has been cut by drill holes too widely spaced to assure continuity.

The 17,220,000 tons of total coal reserve estimated to be amenable for mining is in-situ reserve combining proven and probable reserves. As a general rule, a geological safety factor and mining recovery factor are multiplied into the in-situ reserve to obtain mineable reserve.

The geological safety factor is usually decided by a geologist engaged in the particular geological exploration based on his experience, considering all aspects of the geological conditions. From our experiences, it has been learned that a geological safety factor of around 90–95% is reasonably applied to the area where coal seam occurrence is fairly stable, accordingly, it may be reasonable to adopt a geological safety factor of around 80% for the Unong pit where the geology is considerably complicated. The mining recovery factor is also decided based on the experiences, considering geological conditions, mining method and mining conditions. A verification of the actual mining recovery factor has not been done at the mine site, however, it is estimated to be around 90%.

Applying those factors, mineable reserve is estimated as follows:

 $17,220,000 \times 0.8 \times 0.9 = 12,400,000$

This tonnage is for a whole seam extraction, so-called "run-of-mine". The tonnage of the mineable reserve is reduced further down in case of the currently performed selective mining, since mudstone ply #2, #8, #10 are removed from the Main seam with 10 cm of coal at each contact.

Estimating from the thickness of the seam and each ply indicated in Fig. 13-2, approximately 28% of the volume is removed as waste from the coal seam thickness of about 21 m, i.e., the aggregate thickness of the mudstone plies #2, #8, #10 and coal to be removed at each contract are 5 m and 0.8 m respectively. Therefore, the original mineable reserve for the selective mining is estimated as follows, taking into account the volumetric ratio of the Main seam and Minor seam.

 $12,400,000 \ge 0.81 \ge (1-0.28) + 12,400,000 \ge 0.19 = 9,587,680$ = 9,600,000 tons

16-2 Remaining Reserves (as of the end of 1986)

1) Remaining Reserves on the Basis of Austromineral Estimation

The remaining reserves as of the end of 1986 are estimated at 7,975,000 tons for selective mining by subtracting the mined out reserves by the end of 1986.

 Original in-situ reserves for "run-of-mine" 17,220. Original mineable reserves for "run-of-mine" 12,400. 17,220 x 0.8 x 0.9 = 12,400 Mined out reserves by "run-of-mine" 566 Production from 1979 to 1984 (ROM) Remaining mineable reserve as of the end of 1984 for "run-of-mine" 12,400 - 566 = 11,834 Reserve ratio of the Main seam and the Minor seam 81 : 19 Remaining mineable reserves as of the end of 1984 for 9,150 Selective mining" 11,834 x 0.81 x (1 - 0.28) = 9,150 Mined out reserve from 1986 to 1986 by "selective mining" 1,175 	Reserves	Tonnage x 1,000
 Original mineable reserves for "run-of-mine" 12,400 17,220 x 0.8 x 0.9 = 12,400 Mined out reserves by "run-of-mine" 566 Production from 1979 to 1984 (ROM) Remaining mineable reserve as of the end of 1984 for "run-of-mine" 12,400 - 566 = 11,834 Reserve ratio of the Main seam and the Minor seam 81 : 19 Remaining mineable reserves as of the end of 1984 for 9,150 Remaining mineable reserves for the end of 1984 for 9,150 Mined out reserve from 1986 to 1986 by "selective mining" 1,175 	o Original in-situ reserves for "run-of-mine"	17,220
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 "run-of-mine" 12,400 - 566 = 11,834 Reserve ratio of the Main seam and the Minor seam 81 : 19 Remaining mineable reserves as of the end of 1984 for "selective mining" 11,834 x 0.19 + 11,834 x 0.81 x (1 - 0.28) = 9,150 Mined out reserve from 1986 to 1986 by "selective mining" 1,175 	Production from 1979 to 1984 (ROM)	n na se
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 Reserve ratio of the Main seam and the Minor seam 81:19 Remaining mineable reserves as of the end of 1984 for 9,150 "selective mining" 11,834 x 0.19 + 11,834 x 0.81 x (1 - 0.28) = 9,150 Mined out reserve from 1986 to 1986 by "selective mining" 1,175 		
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 Remaining mineable reserves as of the end of 1984 for 9,150 "selective mining" 11,834 x 0.19 + 11,834 x 0.81 x (1 - 0.28) = 9,150 Mined out reserve from 1986 to 1986 by "selective mining" 1,175 		
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 "selective mining" 11,834 x 0.19 + 11,834 x 0.81 x (1 - 0.28) = 9,150 Mined out reserve from 1986 to 1986 by "selective mining" 1,175 	an a	
$11,834 \ge 0.19 + 11,834 \ge 0.81 \ge (1 - 0.28) = 9,150$ Mined out reserve from 1986 to 1986 by "selective mining" 1,175	Remaining mineable reserves as of the end of 1984 for	9,150
D Mined out reserve from 1986 to 1986 by "selective mining" 1,175	"selective mining"	
D Mined out reserve from 1986 to 1986 by "selective mining" 1,175	$11,834 \ge 0.19 + 11,834 \ge 0.81 \ge (1 - 0.28) = 9,150$	
	Mined out reserve from 1986 to 1986 by "selective mining"	1,175
Domaining respects as of the and of 1086 for		
	o Remaining reserve as of the end of 1986 for	7,975
"selective mining"		

The details of the calculation is indicated in Table 16-2.

2) Remaining Reserve Evaluation

On the basis of the geological data provided by SCC through NAPOCOR in October 1987, when the survey team was in the Philippines for the presentation of the interim report, preliminary reserve calculation has been performed by using a computer as well as by manual method to verify the above reserves derived from the Austromineral F/S report.

The data consists of 15 geologic sections of NE-SW, 7 geologic sections of

•

NW-SE, geological map, topographical map and slope plan map.

In order to proceed accurate reserve calculation, drill core logging data, seam structure contour map, ultimate pit plan and the like are required, however, those data was not available.

Accordingly, the calculation of the remaining reserves has been done on the Main seam which occupies approximately 81% of the total reserves and the total remaining reserves are estimated by adding coal reserves from the Minor seam which occupies about 19% of the total reserves. Due to the nature of the data base, it may only be a preliminary estimate.

Cross section method

a.

On the basis of the provided geologic cross sections from Nos. 1 to 13 dated July 1986, the reserves for selective mining are calculated by the equations indicated below:

In-situ Main seam volume (Vma)

$$V_{ma} = \sum_{n=0}^{13} (\frac{A_n + A_{n+1}}{2} \times D) - F$$

Where,

 A_n : Area of the Main seam within the ultimate pit limit on section n.

D: Distance between two adjacent sections (88 m)

P: Volume of partings with the thickness of more than 0.3 m (21%). Percentage of the partings, which are mudstone plies, with more than 0.3 m thickness in the Main seam is calculated according to the actual reading indicated along with the drill hole columns on each section.

In-situ Minor seam volume (Vmi)

16-5

 $Vmi = \frac{0.19 (Vma + P)}{(1 - 0.19)}$

In-situ reserves (Ri)

 $Ri = (Vma + Vmi) \times SG$ Where,

SG: Specific gravity (1.3) Mineable reserves (Rm) Rm = Ri x GF x MF

Where,

GF: Geological safety factor (0.8) MF: Mining recovery factor (0.9)

Strip ratio

Strip ratio (SR) is calculated by the following equation:

$$SR = \frac{Vt - Vma - Vmi}{Rm}$$

Where,

Vt: Total volume including both coal and waste in the ultimate pit

The volume between the cross section 2 and 13 are calculated by the cross section method and the northwest and southeast areas beyond the section 2 and 13 are calculated roughly based on the Unong pit slope plan or a 1:2,000 scale.

The results of the calculations are summarized as follows:

Mineable coal reserves	7,172,000 tons
Main seam	5,530,000 tons
Minor seams	1,642,000 tons

Waste volume Strip ratio

b.

89,071,000 BCM 12.4 BCM/ton

Reserve calculation by computer

The coal thickness with only less than 0.3 m partings which are unable to remove as waste is figured out from the columnar sections of each boring holes indicated in the cross sections and used as computer input data. Though the indications of the columnal sections are not clear, those are the only data available for the computer operation. The area for the reserve calculation is restricted between cross sections 2.5 to 13, because the locations of the drill holes are not indicated on the geological plan out-side of the said cross sections. The coal reserves in the area which is not covered by the computer calculation are estimated by the manual method.

The results are summarized as follows:

Mineable coal reserves calculated

by computer

Main seam Minor seams 5,970,000 tons 4,600,000 tons 1,370,000 tons

Reserves in the area outside of the cross sections 2.5 to 13

Total estimated mineable reserves

310,000 tons 6,280,000 tons

16-3 Summary

Although there are some differences observed among those three estimates, the remaining reserves in the Unong pit may be in the range of 6 to 7 million tons for selective mining based on the given data and information. It is less than a half of the originally estimated 17,220,000 tons. Table 16-3 shows the summary of the remaining reserve estimates.

As the estimates are done based on the very limited data involving various assumptions, the detail reserve calculation must be implemented based on the more accurate geological data to assure the reserves on which mine production plan is established.

	Reserve Tonnage x 1,000	Remarks
Austromineral F/S basis	7,975	Down to -150 m
andre and the second	an a	Consideration of pit slope is not clear.
	7,172	Down to -140 m based on
(Cross section method)		the SCC cross sections.
Computer	6,280	Down to -140 m based on
a da ser en el compositor de la compositor en la compositor de la compositor de la compositor de la compositor en la compositor de la comp	a Marajara (n. 1997) 1997 - Santa Santa Santa 1997 - Santa Santa Santa Santa Santa Santa Santa Santa Santa Santa 1997 - Santa	including 310,000 tons of
		reserves calculated manually at the northern tip of the pit,
		where data is missing for computer calculation.

Table 16-3Remaining Reserve Estimates at Unong Pit
(for selective mining as of the end of 1986)

Note: The SCC geologic sections show the pit limit down to only -140 m.

The accurate geological data is indispensable for the estimation of accurate and realistic mineable coal reserves, thus the following items by which reserves are seriously affected must be studied precisely.

1) Seam Correlation

It is observed in the both cross sections and the seam thickness isopack map drawn by the computer that the thickness of the Main seam and number of partings vary abruptly from hole to hole, it may indicate that the some of those abrupt variations are the results of faultings. In addition, there remain some concerns to the continuity and correlation of the minor seams.

2) Geological Structure

On the basis of the provided geological cross sections, structure contour map of the Main seam have been drawn so as to evaluate the current interpretation of the geological structure at the Unong pit. According to the contour map, the structure seems to be fairly complicated due to faultings and foldings in the northwestern area and in some other areas, the continuity of the seam is highly suspected between cross sections in which the coal seams are drawn continuously, though.

It is recommended to review the interpretation of the geological structure by drawing accurate structure contour map along with cross sections.

Pit Slope Angle

3)

4)

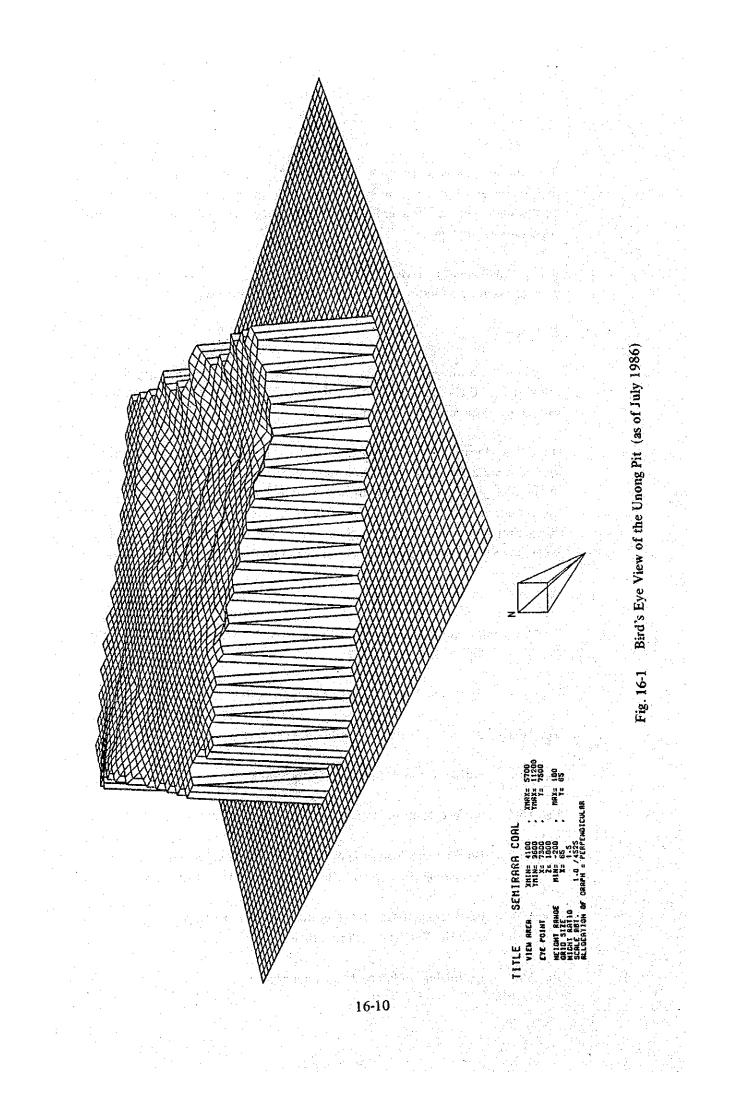
At present, the pit slope of the Unong pit is designed at 1: 3.5 (16°), 1: 3.0 (18.4°), 1: 2.0 (26.6°) and 1: 4.5 - 4.8 (12.5° - 11.8°) for the slopes on the east, south, west and northeast respectively.

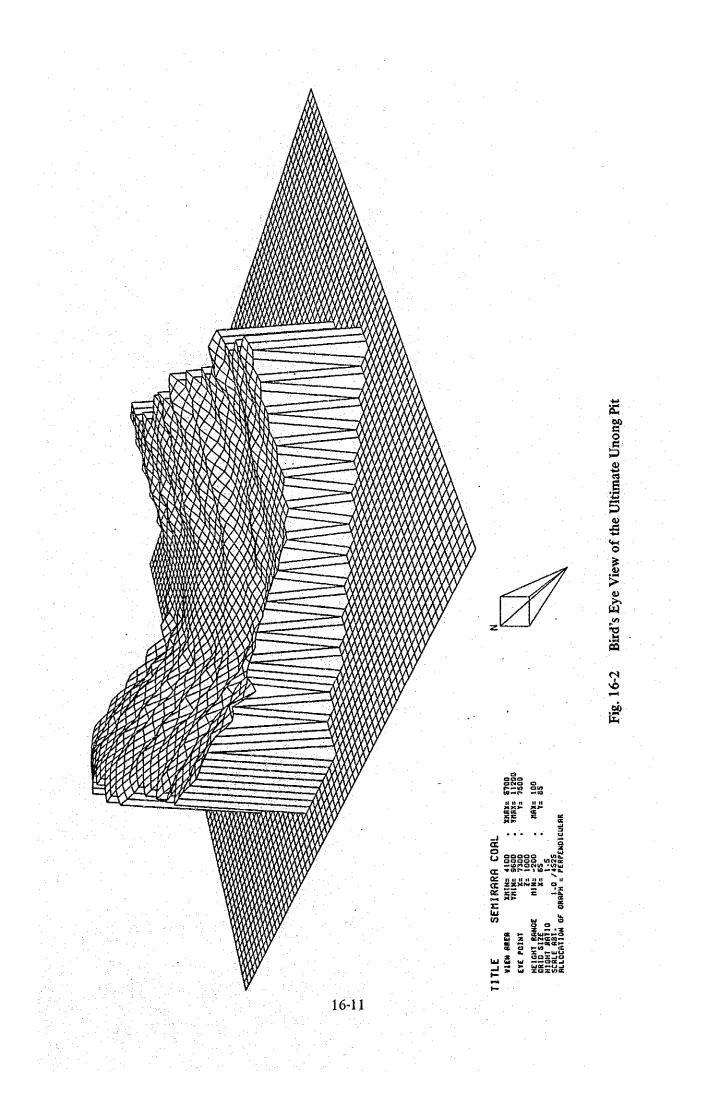
On the northeastern slope, partial slides have been experienced. Presently, the slides are holding their positions stopping the further movement, according to the SCC geologists who are in charge of the slope monitoring, but if the slope angle has to be reduced to avoid the progression of the slides, the pit excavation would be unable to go down to the originally designed 150 m below sea level. Consequently, the mineable reserves would be reduced further.

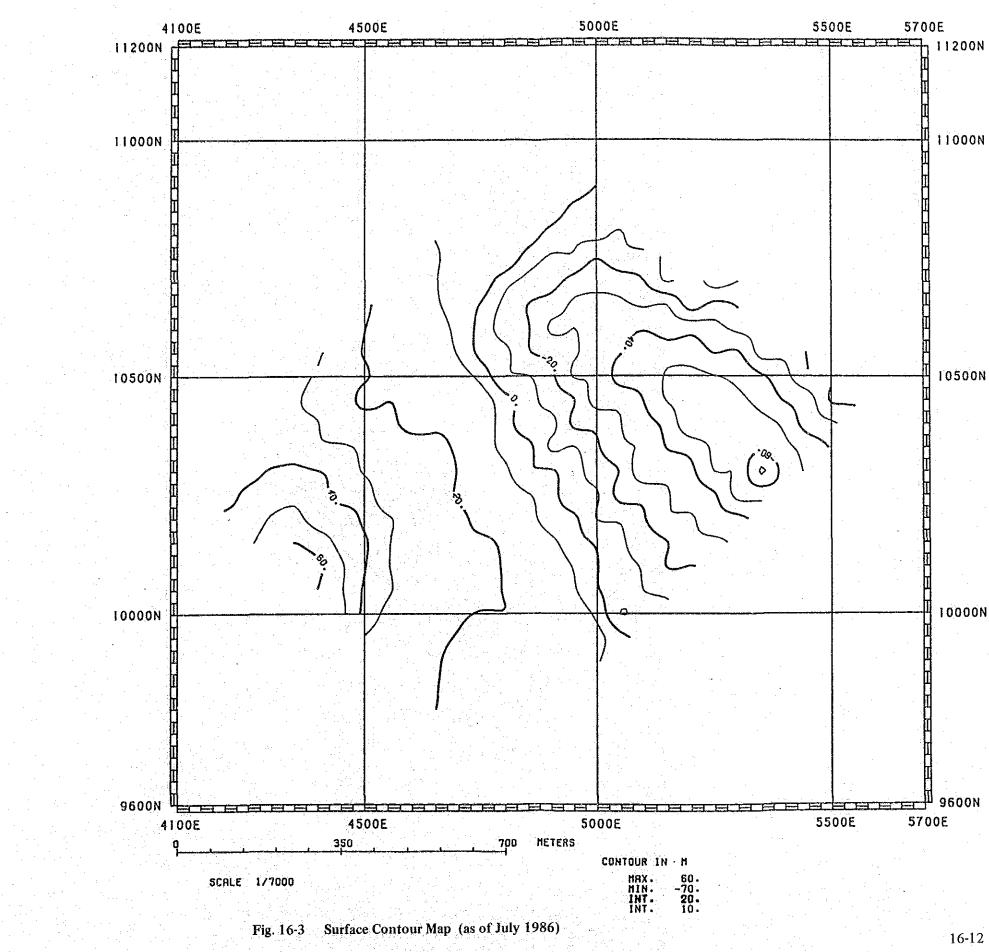
Annexes

The following figures are drawn by the computer system on the basis of the provided geological data.

- Fig. 16-1 Bird's Eye View of the Unong Pit (as of July 1986)
- Fig. 16-2 Bird's Eye View of the Ultimate Unong Pit
- Fig. 16-3 Surface Contour Map (as of July 1986)
- Fig. 16-4 Surface Contour Map of the Ultimate Unong Pit
- Fig. 16-5 Iso-thickness Map of Main Seam for "Selective Mining" (excluding partings with the thickness of more than 0.3 m)
 - Fig. 16-6 Iso-thickness Map of Aggregate Main Seam Partings with the Thickness of more than 0.3 m
- Fig. 16-7 Iso-thickness Map of Total Main Seam



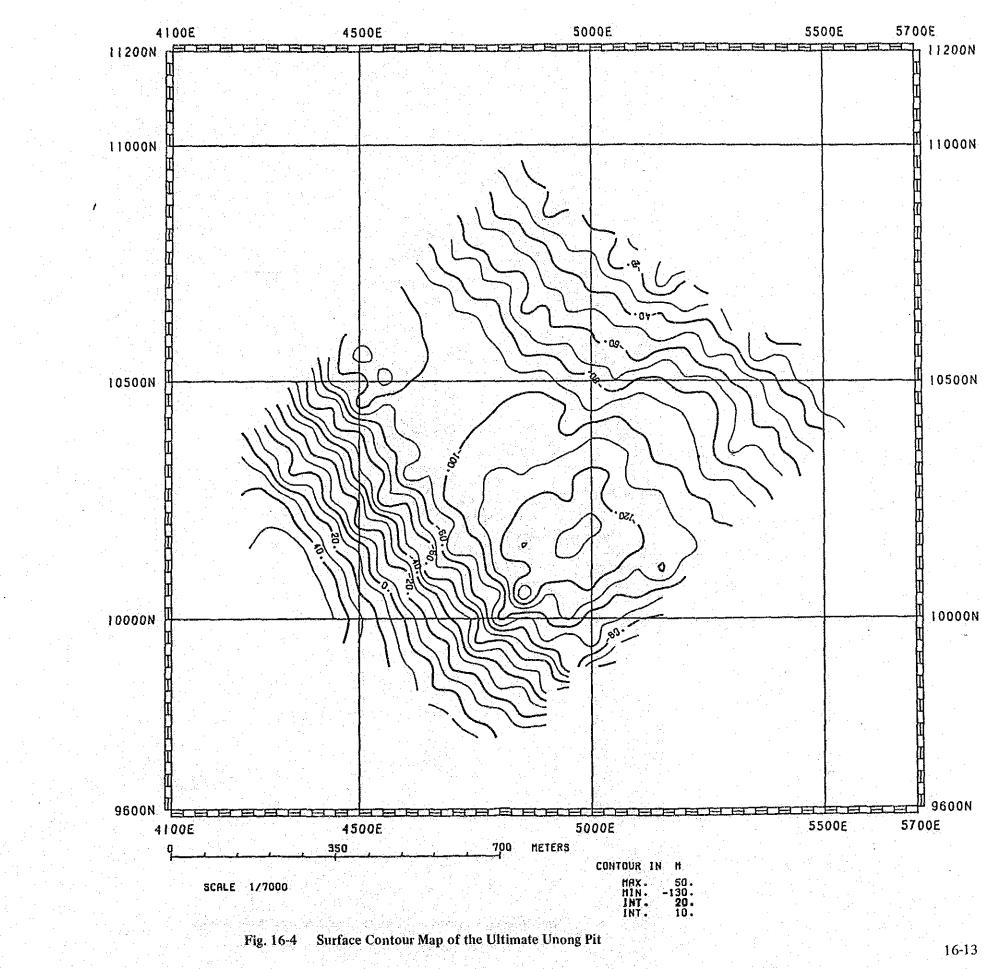




11000N

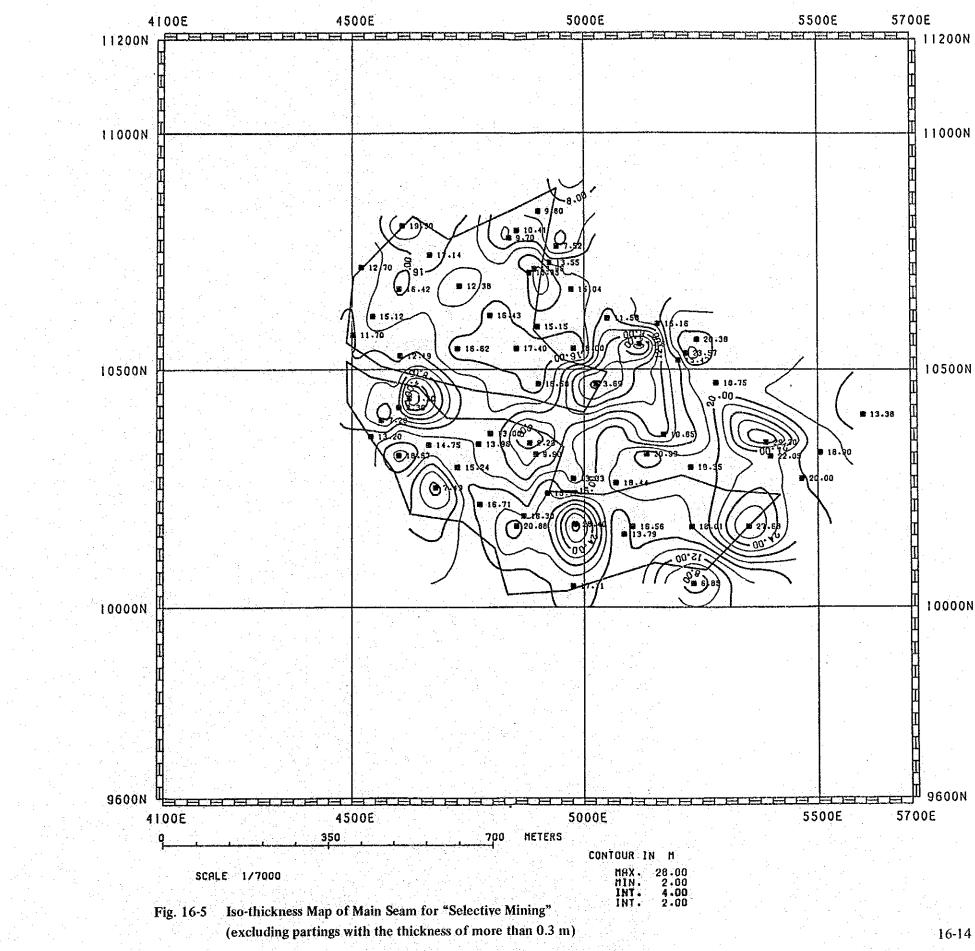
10500N

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4 11000N

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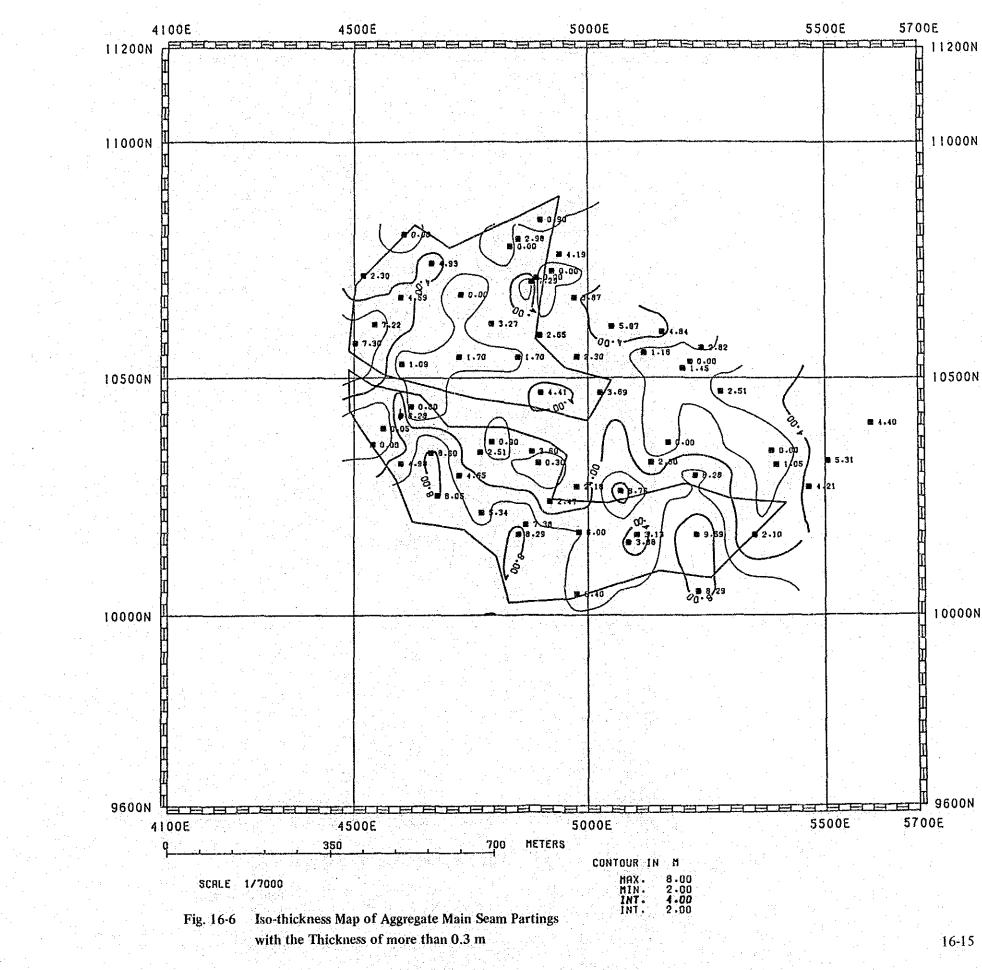


11000N

10500N

10000N

=___ 9600N

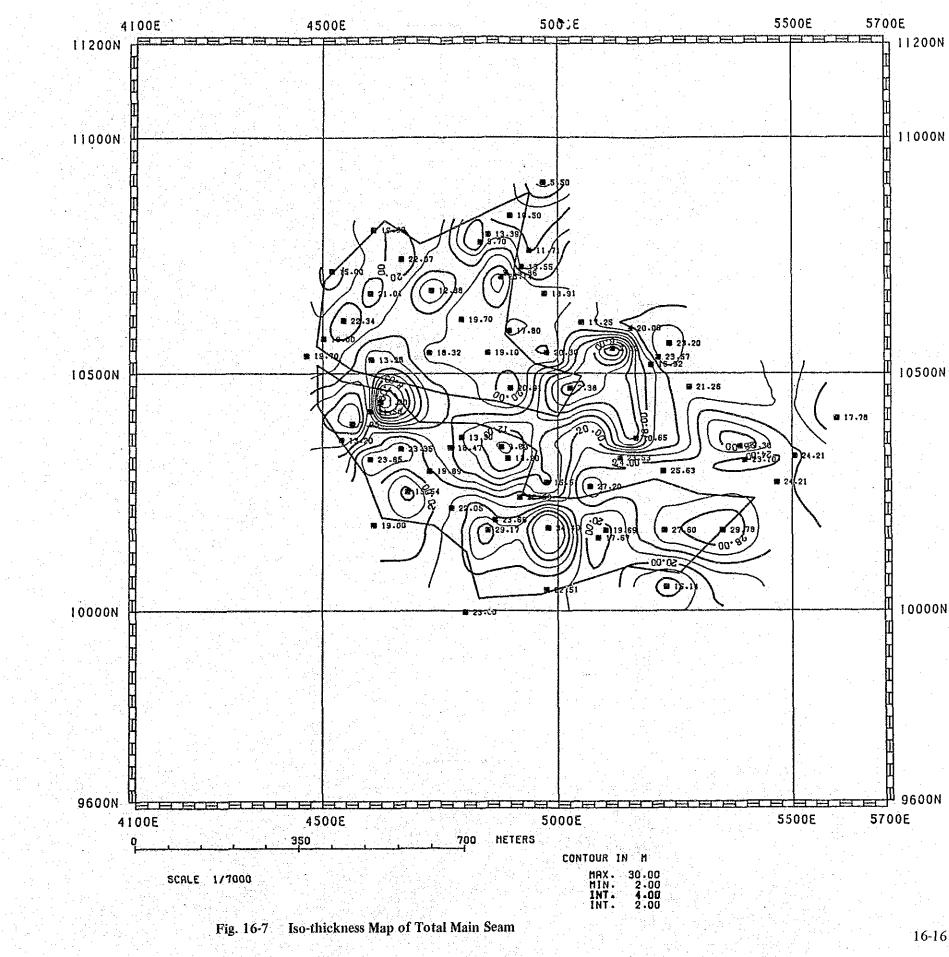


11000N

10500N

10000N

<u>∈</u> ₩ 9600N



4 11000N

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10000N

17. Production Increasing Schemes

- 17-1 Unong Pit Production Plan
- 17-2 Production Comparison Plan vs. Actual
- 17-3 Production Increase Schemes
- 17-4 Summary

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17. Production Increase Schemes

17-1 Unong Pit Production Plan

At present, the long term production plan is available from the feasibility study prepared by Austromineral. Although some uncertainty remains in the accuracy of the study, it might be considered to indicate the general guide of the future mining.

Table 17-1 shows the original mine production plan by Austromineral.

ASTRO	OVBDN (BCM)	COAL (TONS)	TOTAL BCM	S/R
1980	403	60	449	6.7
1981	784	100	861	7.8
1982	1,693	200	1,846	8.5
1983	7,887	300	8,117	26.3
1984	9,257	1,000	10,032	9.3
1985	11,370	1,000	12,054	11.4
1986	12,445	1,000	13,229	12.4
1987	10,565	1,000	11,339	10.6
1988	10,348	1,000	11,122	10.3
1989	10,710	1,000	11,484	10.7
1990	11,379	1,000	12,153	11.4
1991	10,310	1,000	11,084	10.3
1992	7,710	1,000	8,485	7.7
1993	3,210	1,000	3,984	3.2
1994	2,770	1,000	3,544	2.8
1995	3,710	1,000	4,484	3.7
1996	4,190	1,000	4,960	4.2
1997	3,670	1,000	4,445	3.7
1998	3 350	1,000	4,125	3.4
1999	1,540	1,000	2,315	1.5
2000	730	628	1,213	1.2

 Table 17-1
 Mine Production Plan (by Austromineral)

17-2 Production Comparison Plan vs. Actual

Since the production commencement in 1979, below listed production has been done at the Unong pit by the end of 1986.

Waste	25,615,000 BCM
Coal	1,677,000 BCM
	2,206,000 tons
Total BCM	27,295,000 BCM
Stripping Ratio	11.6

Table 17-2 shows a comparison between the production plan and actual.

According to the production plan, approximately 2,152,000 tons of coal production is expected by removing 25,615,000 BCM of waste which is the actual production achieved by the end of 1986. The actual coal release is 2,206,000 tons which is 54,000 tons higher than the above mentioned planned tonnage. As the difference is not so significant for a mine production, it could be considered that the mining operation has been appropriately following the mine plan provided that the actual mining blocks were same as the mining plan.

Table 17-3 shows the mine production comparison between the plan and the actual until 1986. Fig. 17-1 to 17-3 graphically indicate the correlation of the plan and the actual.

	Mine plan up to 1984	Mine plan up to 1985	Actual up to 1986	Estimated production plan based on actual overburden removal BCM	Variance actual minus estimated plan
		· · · ·			
Overburden BCM	20,024,000	31,394,000	25,616,977	25,616,977	0
Coal BCM	1,281,000	1,965,000	1,678,423	1,617,464	60,959
Tons	1,660,000	2,660,000	2,205,594	2,151,907	53,687
TOUS	1,000,000	2,000,000			
Total BCM	21,305,000	33,359,000	27,295,400	27,234,441	60,959
Stripping ratio	12.1	11.8	11.6	11.9	-0,3

Table 17-2 Mine Production Comparison Plan vs. Actual

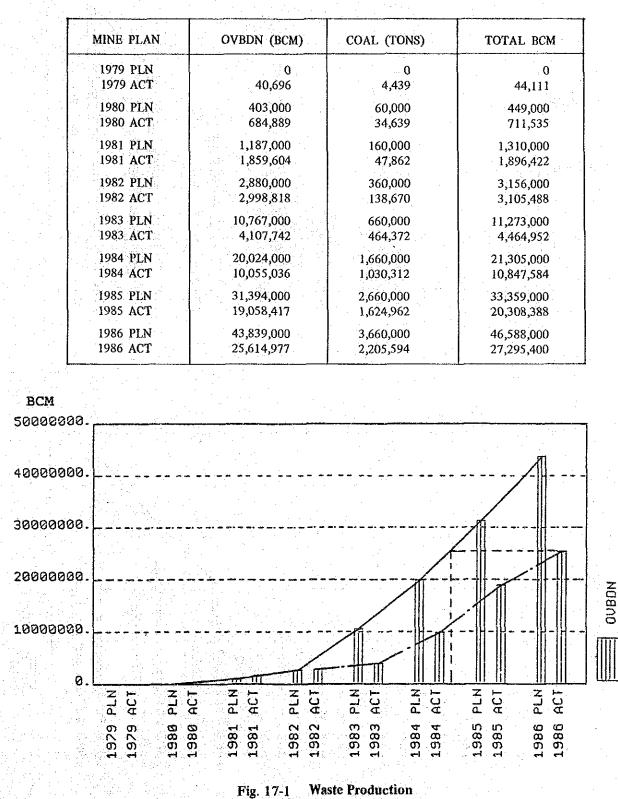
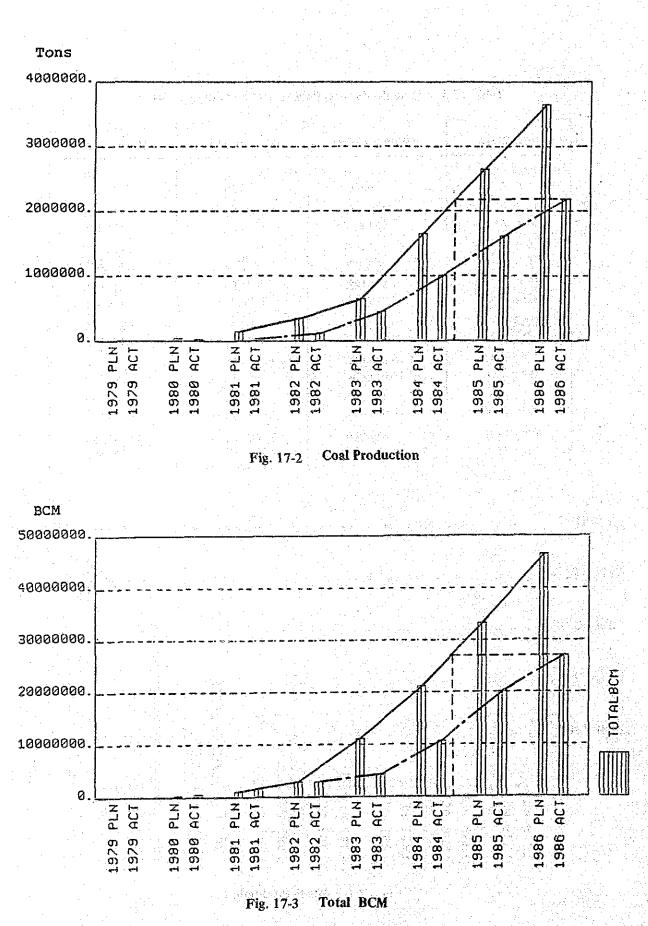


Table 17-3 Mine Production Paln vs. Actual (Cumulative)



17-3 Production Increase Schemes

Several schemes are conceived to increase the current coal production at the Unong pit as mentioned below.

1) Introduction of Larger Capacity Bucket Wheel Excavators

This scheme is not practical, since it requires a replacement of all presently installed BWE system, including belt conveyor system, to a larger capacity system, resulting in double investment of about \$100 million. Furthermore, the system replacement causes a halt of mine production. Considering the remaining very limited coal reserve, estimated to be 6-7 million tons, in the Unong pit this scheme is far from the acceptable range.

2) Introduction of Additional Bucket Wheel Excavator Unit

This scheme is not economically acceptable since the additional BWE unit and another belt conveyor system with spreader requires roughly \$25 million. Considering the physical size and complication of the pit, it would create congestion in the pit. As the remaining reserve is very scarce, it is not worth to introduce additional BWE system, expending large capital cost.

3) Introduction of Additional Conventional Mining Equipment

It is commonly known from the past experience that conventional mining equipment, which is trucks and shovels, is not efficiently used in the pit operation during the wet season when more than 2,000 mm of rainfall occurs. The utilization, which is a percentage of operating hour against available hours, is estimated to be around 60%. Weather delay is estimated at around 30% of the total hours. Considering the high weather delay percentage, it is not advisable to introduce additional trucks and shovels for mine production improvement, however, as only for reference, the system performance is estimated for one front shovel operation.

Table 17-4 shows shovel performance estimate and Table 17-5 for required number of trucks.

The system is a combination of a shovel with 3.1 m^3 bucket capacity and 35 tons (US tons) payload capacity truck.

Table 17-4 Shovel Production Estimate

	117	and the second	1. A.	
Material	Waste			
	S.G.	2.2 tons/BCM	an an Angeler an an An Angeler	
		1.8 tons/LCM		
	Swell	22 %		an an the
		n an Allandar († 1947) National († 1947) - Allandar († 1947) - Allandar († 1947)		
Front shovel				· · ·
	Bucket capa.	3.1 cu. m		
	Bucket fill	0.9		
	Physical avil	75 %	all i shakara ay ad	
	Cycle time	2.3 min.		
	1st dump	0.05 min.	en Menore internet i des pro	dig e se
	2nd pass	0.45 min.		
	3rd pass	0.45 min.		
•	4th pass	0.45 min.		
	5th pass	0.45 min.		n na standar († 1947) 1970 - Angelei Angelei († 1947) 1970 - Angelei Angelei († 1947)
	6th pass	0.45 min.		
			golf and state there	
	Total	2.3 min.		
	Production/hr Actual bu	cket capacity		2.29
	Production/hr Actual bu 3.1*.9*(1/	cket capacity /(1+.22))		2.29
	Production/hr Actual bud 3.1*.9*(1) Loading p	cket capacity /(1+.22)) asses to 35 tons trucks		
	Production/hr Actual bud 3.1*.9*(1, Loading p Truck load	cket capacity /(1+.22)) asses to 35 tons trucks d BCM		6 13.72
	Production/hr Actual bud 3.1*.9*(1, Loading p Truck load	cket capacity /(1+.22)) asses to 35 tons trucks		6
	Production/hr Actual bud 3.1*.9*(1) Loading p Truck load Hourly pro	cket capacity /(1+.22)) asses to 35 tons trucks d BCM		6 13.72
	Production/hr Actual bud 3.1*.9*(1) Loading p Truck load Hourly pro	cket capacity /(1+.22)) asses to 35 tons trucks d BCM oduction BCM/hr aduction BCM/yr		6 13.72
Total days	Production/hr Actual buo 3.1*.9*(1, Loading p Truck load Hourly pro	cket capacity /(1+.22)) asses to 35 tons trucks d BCM oduction BCM/hr aduction BCM/yr	301	6 13.72
Maint. days	Production/hr Actual buo 3.1*.9*(1, Loading p Truck load Hourly pro	cket capacity /(1+.22)) masses to 35 tons trucks d BCM oduction BCM/hr oduction BCM/yr 249 62.3	301 75.3	6 13.72
Maint. days Available days	Production/hr Actual bud 3.1*.9*(1/ Loading p Truck load Hourly pro Yearl shovel pro	cket capacity /(1+.22)) masses to 35 tons trucks d BCM oduction BCM/hr pduction BCM/yr 249 62.3 186.8	301 75 3 225.8	6 13.72
Maint. days Available days Weather delay days	Production/hr Actual bud 3.1*.9*(1/ Loading p Truck load Hourly pro Yearl shovel pro	cket capacity /(1+.22)) masses to 35 tons trucks d BCM oduction BCM/hr oduction BCM/yr 249 62.3 186.8 74.7	301 75.3 225.8 90.3	6 13.72
Maint. days Available days Weather delay days Operating days	Production/hr Actual bud 3.1*.9*(1/ Loading p Truck load Hourly pro Yearl shovel pro	cket capacity /(1+.22)) masses to 35 tons trucks d BCM oduction BCM/hr oduction BCM/yr 249 62.3 186.8 74.7 112.1	301 75 3 225.8 90.3 135.5	6 13.72
Maint. days Available days Weather delay days Operating days Working days	Production/hr Actual bud 3.1*.9*(1/ Loading p Truck load Hourly pro Yearl shovel pro	cket capacity /(1+.22)) masses to 35 tons trucks d BCM oduction BCM/hr eduction BCM/yr 249 62.3 186.8 74.7 112.1 89.6	301 75.3 225.8 90.3 135.5 108.4	6 13.72
Maint. days Available days Weather delay days Operating days	Production/hr Actual bud 3.1*.9*(1/ Loading p Truck load Hourly pro Yearl shovel pro	cket capacity /(1+.22)) masses to 35 tons trucks d BCM oduction BCM/hr oduction BCM/yr 249 62.3 186.8 74.7 112.1	301 75 3 225.8 90.3 135.5	6 13.72

17-6

360 90.0 270.0 108 0 162.0 129.6 3,110

185,560

14 A 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4	A REAL PROPERTY.			
Table	17-5	Truck	Fleet	

Assumption:	3 km flat haul (pit to du	mn)	
rosamperon	5 km nat nati (pit to du	mp)	
Time estimate:			
Truck loading	2.3 min,	n The state of the	
Maneuver in load area	0.8 min.		
Maneuver at dump area	0.8 min.		
Dumping	1.2 min.		
Hauling	11.8 min.		
Returning	<u>3.9 min.</u>		
Cycle time	20.8 min.	· · · ·	
Cycle anne	20.0 mm.		
Truck load:	13.72 BCM/Truck		. ¹ .
Hourly production:	39.58 BCM/hr./unit		
Req'd. truck fleet			
Total days/yr.	249	301	360
Working hr./yr.	2,151	2,601	3,110
Req'd. BCM/yr.	128,346	155,149	185,560
BCM/yr./truck	85,130	102,940	123,084
Net req'd. fleet	1.51	1.51	1.51
Truck availability %	76.0	76.0	76.0
Req'd. No. of fleet	1.98	1.98	1.98
No. of trucks	2	2	2

Table 17-6 shows estimated mine production at various stripping ratio. In the next 5 to 6 years, average stripping ratio is anywhere around 1:10 to 1:13 for run-of-mine extraction and 1:14 to 1:18 for selective mining based on the Austromineral mine plan. Thus, the expected production increase per shovel is as indicated in the Table 17-6.

Required capital cost for one shovel and two trucks is approximately \$1.3 million. In addition to those, 2 bulldozers, one for waste dump site and one for shovel assistance, are required, which cost about \$1.2 million.

The limited pit size and presently installed BWE system may restrict the number of shovel unit to be introduced in the pit.

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Table 17-6 Production Estimate

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TOTAL DAYS/YEAR	249		301		360	
YEARL TOTAL BCH	128,346	an a	155, 149		185,560	
	La					
STRIPPING RATIO	COAL TONS	VASTE BCH		VASTE BCH		WASTE BCH
5.0	22,247	106,099	26,892	128,257	32, 164	153, 396
5.5	20,472	107,874		130,401	29,599	155,961
6.0	18,960	109,386	22,920			158,148
6.5	17,656	110,690	21,343		25,527	160,033
7.0	16,520	111,826	19,970	135,179		
. 7.5		112,825				
8.0	14.636			137,457	21,160	
8.5	13,846	114,500	16.738			165,541
9.0	13,138	115,208				166,566
9.5	12, 498	115,848		140,041	18,070	
10.0	11,918	116,428	14,407	140,742	17,231	
10.5	11,389	116,957	13,767	141,382		
11.0	10,905	117,441	13,183	141,966	15,767	169,793
11.5	10,461	117,885	12,645		15,124	170,436
¥ 12.0	10,051	118,295	12,150	142.999	14,532	171,028
A CONTRACTOR OF	9,672	118,674	11,692	143,457	13,984	
13.0	9,321	119,025	11,268			172,084
13.5	8,995	119,351	10,873			172,556
· 14.0	8,690	119,656	- 10,505			172,996
14.5	8,406	119,940	10,161	144,988	12,153	173,407
15.0	8,139	120,207	9,839	145,310	11,767	173,793
15.5		120,457	9,536	145,613	11,406	
a) 15.5 15.5 15.6 16.0 16.5	7,654	120,692	9,252	145,897	11,066	174,494
16.0 16.5 11 16.5	7,432	120,914	8,984	146.165	10,745	
Win 12.0	7,223	121,123	8,731	146,418	10,443	175,117
11.5	7.025	121,321		146,657	10,157	175,403
18.0	6,838	121,508	8,266	146,883	9,886	<u>175,674</u>

4) Increase of Operating Days

Assumptions

Mine production is estimated in various cases based on the geological conditions and coal-waste ratio estimated in the mine plan prepared by Austromineral.

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The assumptions are summarized as follows:

a) a

a.

Existing conventional mining equipment is exclusively used for pit affiliated work such as highwall finishing, sump construction and the like.

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- b) An existence of the hard sandstone is 20% of the total waste volume, which is unable to excavate without prior ripping by bulldozers and/or blasting.
 - Run-of-MineSelective MiningSoft material (1)720Hard material (2)601Coal748 (3)
- c) Cutting rate of BWE's are as listed below:

Note: (1), (2): 1985–1986, actual

(3): 1984, July-December, actual

(4): 1985-1986, actual

- Approximately 28% of the main seam is discarded as waste in selective mining, total seam thickness -21 m, aggregate thickness of mudstone plies #2, #8, #10-5 m, coal loss at each ply contact -0.8 m excluding between #10 and #11.
- Actual percentage of BCM's cutting time is 49.6% derived from the actual performance from 1985 to 1986 and 54.8% from July 1984 to December 1986, except CCP delay estimated only 0.4% which is the actual in 1986 and is considered to be stabilized at the low level.
- f) Three annual operating days are assumed, 249 days for 5 days per week, 301 days for 6 days per week and 360 days for the year-round operation.
- g) Remaining coal reserve in the Unong pit, as of the end of 1986, is shown in Table 17-7 on the basis of Austromineral mining plan.
- h) Maximum yearly production is assumed as listed in Table 17-8 considering the operating time and capacity of the BWE system.

17-9

d)

e)

ASTRO YEAR	OVERBURDEN (x1,000 BCM)	SOFT MAT'L (x1,000 BCM)	HARD MAT [°] L (x1,000 BCM)	COAL TONS (x1,000)	STRIP RATIO
1985	5,777	4,622	1,155	454	12.7
1986	12,445	9,956	2,489	1,000	12,4
1987	10,565	8,452	2,113	1,000	10.6
1988	10,348	8,278	2,070	1,000	10.3
1989	10,710	8,568	2,142	1,000	10.7
1990	11,379	9,103	2,276	1,000	11.4
1991	10,310	8,248	2,062	1,000	10.3
1992	7,710	6,168	1,542	1,000	7.7
1993	3,210	2,568	642	1,000	3.2
1994	2,770	2,216	554	1,000	2.8
1995	3,710	2,968	742	1,000	3.7
1996	4,190	3,352	838	1,000	4.2
1997	3,670	2,936	734	1,000	3.7
1998	3,350	2,680	670	1,000	3.4
1999	1,540	1,232	308	1,000	1.5
2000	730	584	146	628	1.2
TOTAL	102,414	81,931	20,483	15,082	6.8

 Table 17-7
 Remaining Reserve as of the end of 1986

Table 17-8 Maximum Yearly Production Estimate

Yearly	BWE Cutting	Max, Yearly Produ	
Operating	Time	(x 1,000 tons	
Days	(%)	Selective Mining	
249	49.6	1,000	1,400
	54.8	1,100	1,500
301	49.6	1,200	1,600
	54.8	1,300	1,800
360	49.6	1,450	2,000
	54.8	1,600	2,200

13.3

Overland coal conveyor capacity

b.

The nominal capacity of the overland coal coneyor, K1 and K2, is 1,000 tons per hour (1,020 tons per hour to be precise).

Assuming that coal is loaded onto the conveyor at the rate of 1,000 tons per hour for 49.6% or 54.8% of the total hour, which are the estimated percentage of BWE cutting time, approximately 3 to 4.5 million tons of coal, depending upon the operating days, can be transported through the conveyor system per year.

In the run-of-mine extraction, the average coal output of either one of BWE's is 748 tons per hour which is a loading rate to the conveyor system. The annual transportation capacity, in this case, is approximately 2 to 3.5 million tons.

In the selective mining operation, the estimated cutting rate of the coal seam is 599 tons per hour and only 72% of the coal seam cutting time is available for the product coal loading, since the 28% of the whole seam is discarded as waste. In this case, the estimated product coal tonnage to be transported through the conveyor system is approximately 1.5 to 2 million tons per year.

Those estimate is based on the assumption that only one BWE is loading the conveyor system.

Fig. 17-4 shows the estimated transportation capacity of the conveyor system.

As indicated, the transportation capacity of the conveyor system is well sufficient to handle the maximum production scale of all scenarios, if the mining sequence of the 4 BWE's is well organized to avoid simultaneous coal cutting of plural BWE's.

Million tons

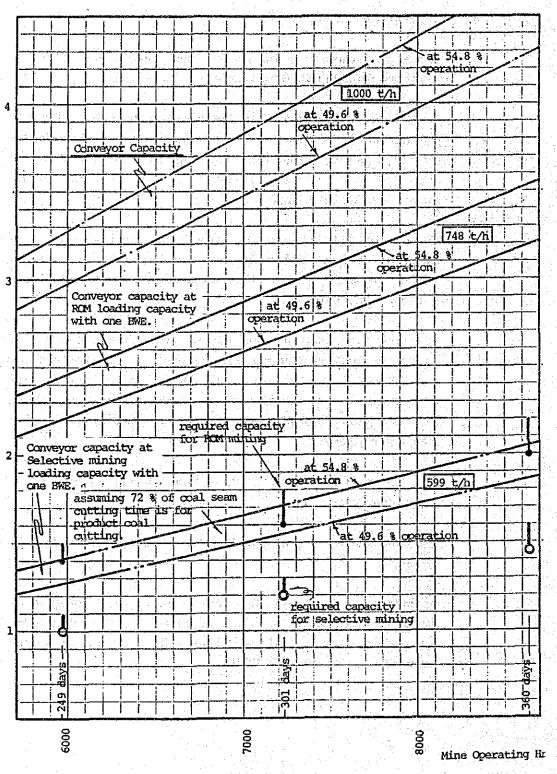


Fig. 17-4 Overland Coal Conveyor Capacity

Selective mining

c.

The	mine production plans are developed on the following c	ases:
a)	249 days per year (5 days per week) operation with cutting time	49.6% BWE Table 17-9
b)	301 days per year (6 days per week) operation with cutting time	49.6% BWE Table 17-10
c)	360 days per year (all the year round) operation BWE cutting time	with 49.6% Table 17-11
d)	249 days per year (5 days per week) operation with cutting time	54.8% BWE Table 17-12
e)	301 days per year (6 days per week) operation with cutting time	54.8% BWE Table 17-13
f)	360 days per year (all the year round) operation with cutting time	54.8% BWE Table 17-14

Selective Mining

Cutting Time	49.6%
Hard Material	20.0%
Hard Material	601 BCM/hr.
Soft Material	720 BCM/hr.
Coal	599 tons/hr.

	and dame.	<u></u>							
TOTAL	81, 931 20, 483 102, 414	3, 248 105, 662	10,859	4, 223	8, 353	7,938 7,903 7,745 5,277 5,277 5,277 5,277 5,277 3,166 2,535 114,016	1. 6		
2003	1.617 404 2.021	2.165	480	187 667	369	2,535	4.5	1.4	
2002	2,020 505 2,525	2,705	600	233 833	462	3,166	4.5	1.7	÷
2001	3,367 842 4,208	239	e de la com	389 1,389	769	5,277	4.5	2.8	
2000	3, 367 842 4, 208	4.508		389 1, 389	769	12.217	4.5	2.8	
1999	3,367 842 4,208		جب	389 1,389	269	5.277	4.5	2.8	· .
1998	3,367 842 4,208		1,000	- in a second second	692	5.277	4.5	2.8	-
1997	5,448 1,362 6,810		875	340 340 340 340	673	17,745	8.1	4	•
1996	5,832 1,458 7,290		574	223 797	441	7,903	L 13.0	4	+ +
1995	5,918 1,479 7,397	151 7,548	500	197 703	389		14.9	4	
1994	5,963 1,491 7,454	141	471	183 654	362	7,957	16.1	4	: :
1993	5,938 1,485 7,423	147	4 30	191 681	377	7,947	15.4	4	
1992	5,922 1,481 7,403	151 7,554	503	196 699	337	31 7,932 7,941	15.0	4	
1991	5,904 1,476 7,380			201	397	7,932	0 14.6	4	
1990	5,914 1,479 7,393	151	504	196 700	Ř	17,931	15.	-	
1989	5,956 1,480 7,445	142	474	184 658	364	7,951	10.0	4	
1988	6.020 6.011 5.956 5.9 1.505 1.503 1.489 1.4 7.525 7.514 7.445 7.3	131	436	170 606	336	7,983 7,980 7,951 7,95	17.8 17.5 16.0	4	
1987	6,020 1,505 7,525	7,653	429	167 596	330	7,983	17.8	-	
	VASTE (BCH) SOFT MATERIAL ILAND MATERIAL SUB-TOTAL	VASIE CUAL	COAL (TONS) PRODUCT COAL	VASTE COAL TOTAL COAL SEAN	PRODUCT COAL BCH	TUTAL BCM	STRIP RATIO	IU IPHENT BUE UN IT	
	VASTE SOFT SUF SUB	LOTAL	COAL	TOTAL	PRODU	[TUTAL	STRIP	EQUIPMENT BUE UNIT	, f

601 720 599

HARD MATER IAL BCM/HR SOFT MATER IAL BCM/HR COAL TONS/HR

240 49.6 20.0

OPLERATING DAYS CUTTING TIME X NARD HATERIALX

NOTES

Semirara Unong Pit Selective Mining

Table 17-9

	1987	1988	1989	1990	1991	1992	1993	1994	1995	1996	1997	1998	1909	2000	TOTAL
VASTE (BCH) SOFT MATERIAL IARD NATERIAL SUB-TOTAL VASTE COAL TOTAL VASTE	7,275 1,819 9,094 9,250	7,266 1,817 9,083 9,241	7,152 1,788 8,940 9,122	7,140 1,785 8,925 8,925 9,111	7,156 1,789 8,945 182 9,127	7, 186 1, 796 8, 982 9, 158	7.195 1.799 8.994 173 9.167	7,128 1,782 8,910 189 9,099	4,747 1,187 5,934 8,293	4, 747 1, 187 5, 934 6, 293	4,747 1,187 5,934 6,293 6,293	4,747 1,187 5,934 6,293	3,956 989 4,945 5,244	1.488 372 1.860 1.973	81,931 20,483 3,248 102,414 105,662
COAL (TONS) PRODUCT COAL VASTE COAL TOTAL COAL SEAN PRODUCT COAL BEAN	520 520 722 400	527 205 732 405	607 236 843 467	621 621 241 862 862	610 237 847 469	588 588 228 816 452	580 225 805 446	631 246 877 486	1,200 467 1,667 923	1,200 467 1,667 923	1,200 467 1,667 923	1.200 467 1.667 923	1,000 389 1,389 769	376 146 522 289	10,859 4,223 15,082 8,353
TOTAL BCN	9,649 9,646	9,646	9,588	9.588 9.588 9.597	9,597	9,610		9,585	7,216	7,216	7,216	7,216	6,013	2,262	<u>9,613 [9,585] 7,216] 7,216] 7,216] 7,216] 7,216] 5,013] 2,262] 114,016</u>
STR IP RATIO	17.8	17.5	15.0	14.7	15.0	15.6		15.8 14.4	5.2	5.2	5.2	5.2		5.2 5.2	9.7
EQU IPACAT BVE UNIT	4	4		*		7	4	4	3.2	3.2	3.2	3.2	2.6	0.1	
	OPERATING DAYS CUTTING TIME & HARD MATERIALX	DPERATING DAYS CUTTING TIME &	301 49.6 20.0		IARD NA Soft NA Coal	MATER IAL MATER IAL	BCN/HR BCN/HR TONS/HR	601 599 720							

Table 17-10 Semirara Unong Pit Selective Mining

Table 17-11 Semirara Unong Pit

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TOTAL	81,931 20,483 102,414	3,248 105,662	10,859	4,223	15,082	8, 353	0 114,016	9.7		
2001	000	00	0	0	0	0	0	8.1 #DIV/0!	0.0	
1998	1,962 490 2,452	127 2,580	426	166	591	328	2,907	8.1	1.1	•
1997	3,686 922 4.608	239	800	311	1,111	615	5,463	6.1	2.0	
1996	4,608 1,152 5,760	299 6.059	1,000	389	1, 389	769	515 111,468 111,474 111,492 111,623 9,901 9,901 9,901 6,828 5,463 2,907	6.1	2.5	
1995	6,682 1,670 8.352	434	-		2,014	1,115	9,901	6.1	3.6	
1994	6,682 1,670 8,352	434 8,786			2,014	1,115	9,901	1 6.1	3.6	
1993	6,682 1,670 8.352			564	_	1,115	9,901	6.1	3.6	
1992	8,596 2,149 10,745	246	822	320	1,142	632	11,623	15.5 13.4	4	
1991	8.590 2.147 0.737	211 211	706	275	981	543	11,492		4	•
1990	8,550 2,137 10.687	220		286	1,023	567	11,474	14.8	4	~ ~
1989	8,548 2,137 10.685	219	733	285	1,018	564	11,468	14.9	-	360 49 6
1988	8,646 2,162 10,808	198			_	509		16.6	4	ING DAYS
1987	8,700 2,175 10,875	187 11,062 11,	624	242	866	180	11.541 11	17.7	4	OPERATING CHTT ING TI
	VASTE (BCH) SOFT NATERIAL NARD NATERIAL SUB-TOTAL	VASTE COAL TOTAL VASTE	COAL (TONS) PRODUCT COAL	VASTE COAL	TOTAL COAL SEAN	PRODUCT COAL BCN	TOTAL BCH	STRIP RATIO	EQUIPMENT BVE UNIT	NOTES

CUTTING TIME X 49.6 HARD MATERIALX 20.0

601 720 599

IIARD MATER IAL BCM/IR Soft Mater IAL BCM/IR Coal 'Ions/IIR

Selective Mining

Cutting Time	54.8%
Hard Material	20.0%
Hard Material	601 BCM/hr.
Soft Material	720 BCM/hr.
Coal	599 tons/hr.

Coal

ر ۔۔۔						. 1				<u>.</u> .				١		٦		1
TOTAL	81.931	20,483	102,414	3,248	105,662			10.859	4,22,	15,082	8,353		114,016		9.7		: .	
2001	2.066	517	2,583	184	2,767			616	239	855	473		_		4.5		1.6	
2000		923					· · · ·	1,100	428	1.528	846		5,791 5,791 5,791 5,791 3,241		4.5		2 8	
1999	3.693	823	4,616	329	4,945			8	428	528	846		5,791		4.5		2.8	
1998	3.693	923	4,616	329	4,945	-		1,100 1,	428	1,528	846		5,791		4.5		2.8	
1997	3.693	923	4,616	329	4,945			1,100	428	1,528	846		5, 791		. 4.5		2 8	
1996	6.006	1.501	7,507	292	7, 799			212	380	1.357	752		8,551 5,791		8.0		4	,
1995	6.478	1,619	8,097	185	8,282			619	241	860	476		8,759	1977 - 19	13.4		4	ן ין
1994	6. 554	1,638	8, 192	164	8, 356		· · ·	548	213	761	421		8,777 8,759		15.3		4	
1993	6.586	1,647	8,233	156	8,389			522	203	725	402		8,791		16.1		4	•
1992	G. 550.	1.638	8, 188	165	8, 353			551	214	765	424		8,776 8,791		15.2		4	
1991	6.534	1.633	8,167	169	8,336			566	220	786	435		8,772		14.7		P	
1990	6.535	1.634	8, 169	168	8,337			563	219	782	433		8,771		14.8		4	
1989	6.561	1.640	8,201	162	8,363	•		541	211	752	416		8,779		15.4	•	7	
1988	6. 641		8,301	144	8,445	- 1		482	187	669	371		8,819 8,816		17.5		7	*
1987	6.650	1.662	8,312	142	8,454			474	185	659	365		8,819		17.8		¥	•
	VASTE (BCN) Snet Watertai	IIARD NATER IAL	SUB-TOTAL	VASTE COAL	TOTAL VASTE		COAL (TONS)	PRODUCT COAL	VASTE COAL	TOTAL COAL SEAN	PRODUCT COAL BCH		IOTAL BCH		STR IP RATIO		EQUIPMENT Ruf Intit	

601 599

HARD MATERIAL BCM/IR SOFT MATERIAL BCM/IR COAL TONS/IR

249 54.8 20.0

OPERATING DAYS CUTTING TIME X HARD MATERIALX

SALOW

Semirara Unong Pit Selective Mining **Table 17-12**

	tan jan Arti					وسنبسو	,		
TOTAL	81, 931 20, 483	102, 414 3, 248 105, 662	10, 859	4, 223 15, 082	8,353	10.660 10.647 10.602 10.602 10.610 10.627 10.602 10.512 1 6.894 1 6.894 1 6.363 1 5.303 1 3.701 1 114.016	9.7		
1999	2,364 591	2,955 209 3,164	698	596 172	537	3,701	4.5	.	
1998	3, 387		-	्र - मै	692	5,303	4.5	2.1	
1997	4,065	5,081 359 5,440	1,200		923	6,363	4.5	2.6	
1996	4,404 1,101	5, 505 389 5, 894	300	506 1,806	1.000	6,894	4,5	2.8	
1995	4,404 1,101	5, 894	1,300	1,806	1,000	6,894	4.5	2.8	
1994	7,682	9, 603 9, 858	851	331 1,182	655	10,512	11.6	4	601 599
1903	7,901	9,876 203 10,079	680	264 944	523	10,602	14.8	4	BCN/IIR BCN/IIR TONS/IIR
1992	7,964	9, 955 188 10, 143	629	245 874	484	10,627	16.1	4	MATER IAL MATER IAL
1001	7,918	9, 898 199 0, 097	999	259 925	512	10,610	15.2	V	HARD HA SOFT HA COAL
1990	7,894 1,973	883 9,867 201 206 084 10,073	688	267 955	529	10,602	14.6	4	
1989	7.906 1.977	9.883 201 10.084	673	262 935	518	10,602	15.0	4	301 54.8 20.0
1988	8,006 2,001	10,045 10,007 179 179 10,217 10,186 10	208	233 832	461	10,647	17.0	4	OPERATING DAYS CUTTING TIME & HARD MATERIAL&
1087	8,036 8,006 2,009 2,001	10,045	575	224 709	443	10,660	17.8	4	OPERAT I CUTTING IIARD NA
	RF	SUB-TOTAL VASTE COAL TOTAL VASTE	COAL (TONS) PRODUCT COAL	WASTE COAL TOTAL COAL SEAN	PRODUCT COAL BCM	TOTAL BCH	STRIP RATIO	EQUIPHENT BVE UNIT	
	VAS'H SOFT HARE	VASI	COAL	TOTA	PROD	TOTA	STRI	EQUI	NOTES

Table 17-13Semirara Unong PitSelective Mining