

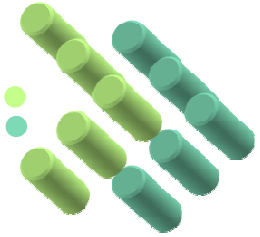
SOMA LIGNITE BASIN, TURKEY

이희욱

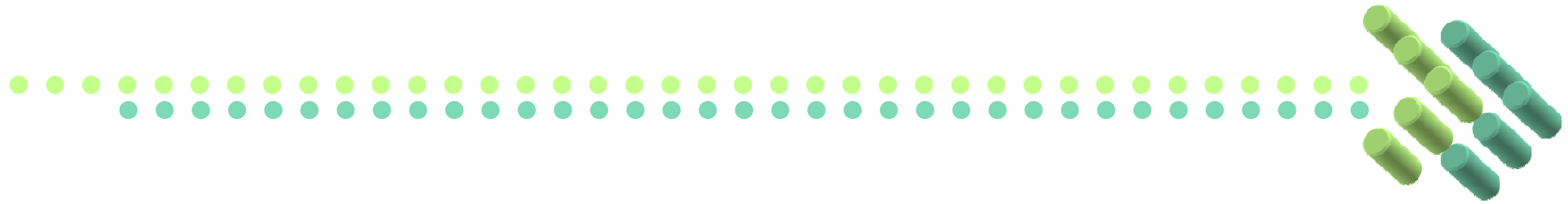
2009. 11. 18.



Contents

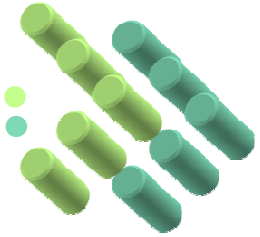


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13.1 Introduction

13.1 Introduction



- **Lignite (갈탄)**

low pressure, low temperature 에서 생성

low rank coal

수분함량이 높음 (max. 66 %)

Ash 성분도 많음 → 건조시키면 가루로 변함

낮은 에너지 밀도 → 운송 수단의 연료로는 부적합

주로 산지와 가까운 화력 발전에서 사용됨

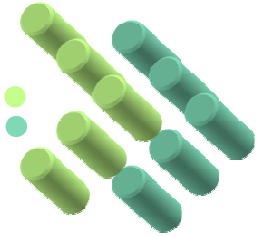


Lignite mine (독일 북서부 노르트라인베스트팔렌 주 퀴른 시)



(emuseum.go.kr)

13.1 Introduction



- **SOMA LIGNITE BASIN** 개발 배경

Energy crisis caused by the OPEC oil price rises which started in 1973

Search for alternative fossil fuels

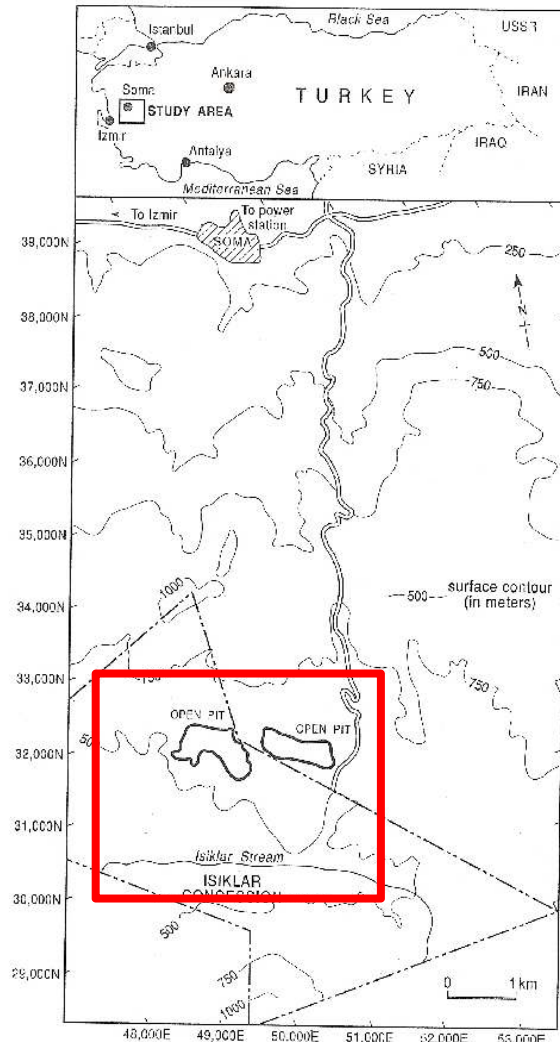
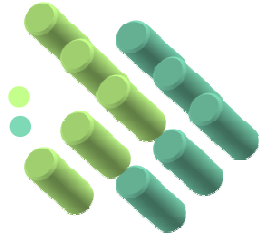
Important for developing countries which wish to reduce their reliance on imported oil

→ **exploitation of Soma Isiklar lignite deposit in western turkey**

→ **feedstock for the Power Station (A, B)**

→ **2 Mt 를 광산 확장을 통해 증가**

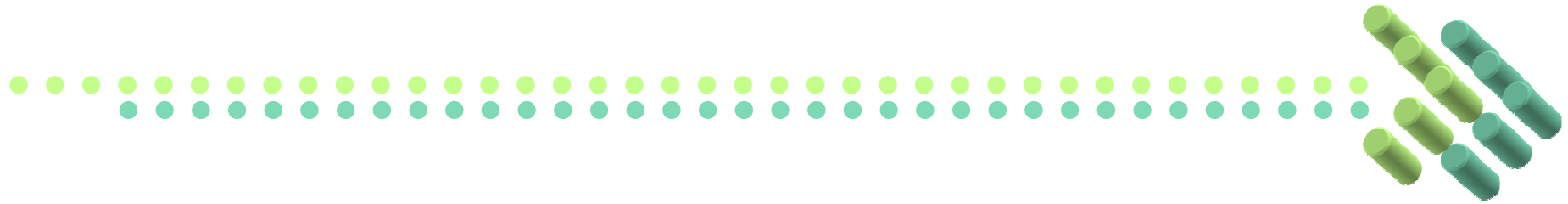
13.1.1 Location



- **Soma lignite deposit**

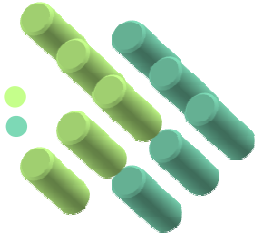
- In the Manisa Province of western turkey
- 10 km south of the town of Soma
- South-facing slope
- Elevations ranging from 750 m in the north to 310 m in the south

(fig. 13.1)



13.2 Exploration programs

13.2.1 Previous work



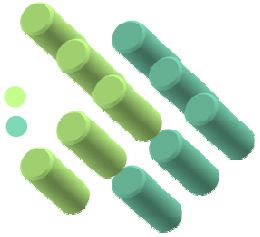
- **Database : four drilling programs**

(table 13.1)

Drilling program	Number of holes	Total meterage	Drilling date
200	34	8790	1960
100	50	10,825	1976
300	9	1547	1981
400	29	8631	1982
Total	122	29,793	

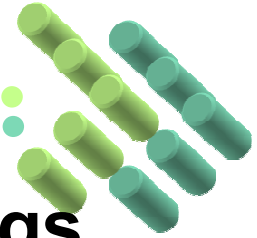
- Analyzed for ash & moisture content, calorific value, sulfur & volatile content
- Produced Isopach, structure contour, isoquality, reserve map
- Geotechnical study, specific gravity

13.2.2 Core recovery



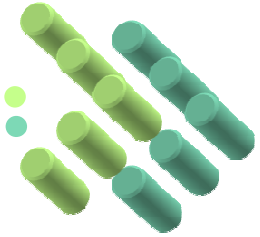
- 코어 회수율이 **95%**를 넘지 못하면 재시추
- 회수율을 증진시키기 위한 방법
 - Use of large diameter wireline
 - Use of air flush core barrels
 - Use of triple tube core barrels

13.2.3 Geophysical logging



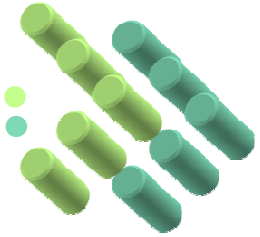
- **Typical DTH(Down-the-hole) geophysical logs**
 - natural gamma, density, neutron, caliper, resistivity, etc
 - Shale, mudstone, marl → high γ , density
 - Coal → low γ , density
 - Neutron → estimating porosity
 - Resistivity → indicate bed boundaries

13.2.4 Sampling

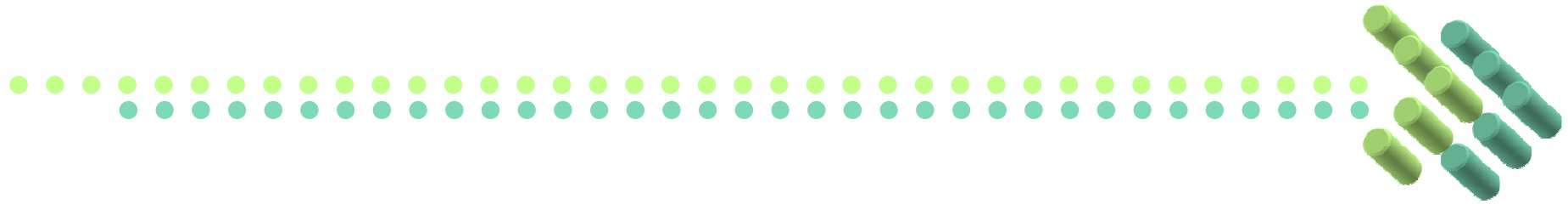


- **Whole lignite sequence was sampled, including all parting material**
- **All lithological layers in the seam greater than 30 cm thick were sampled**
- **Minimum thickness of parting that could be mined as waste in the open pit**

13.2.5 Grouting

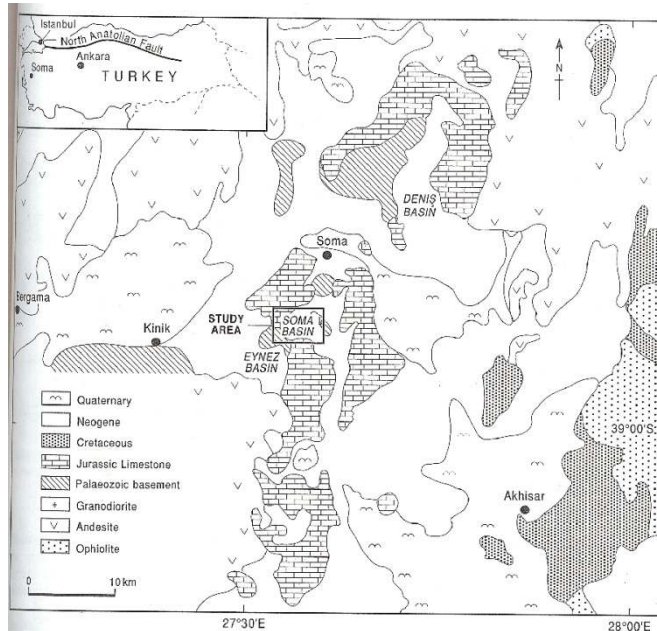
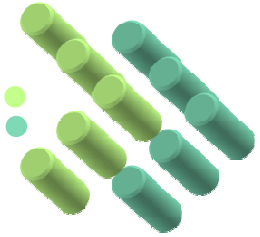


- All holes where deep coal is likely to be mined by **underground methods** drilled prior to mining should be sealed using **pressure grouting**
- reduce potential water Inrush hazard



13.3 Geology

13.3.1 Geological setting

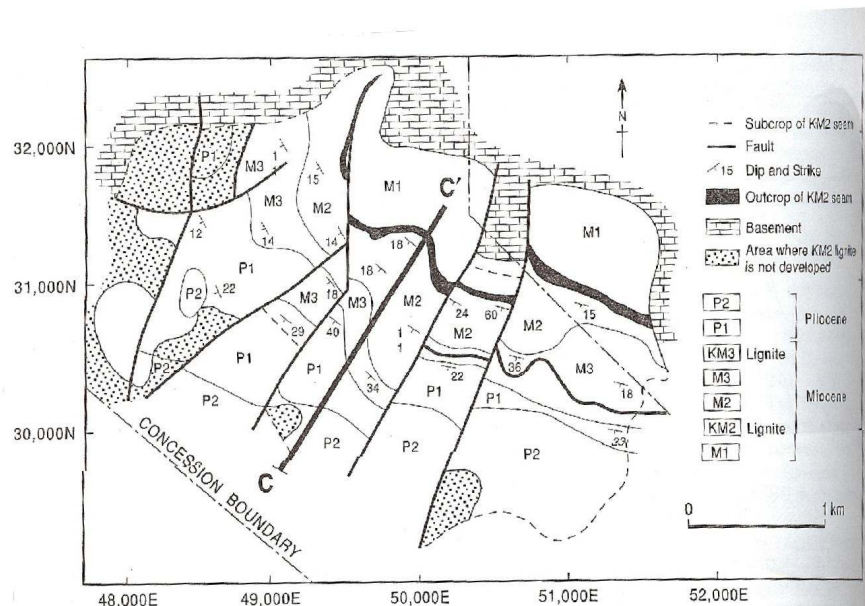
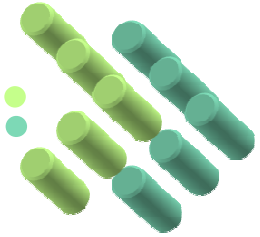


(fig. 13.2)

- Precambrian, Palaeozoic and Mesozoic Sedimentary and igneous rock
- North Anatolian Fault (NAF) : late Serravallian (mid. miocene)
- Series of NE-trending graben*s

* Graben : 단층에 수반하여 상반이 하강하여 생긴 좁고 긴 열곡

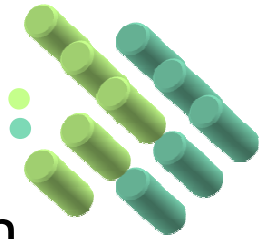
13.3.2 Geology of the Soma Basin



(fig. 13.3)

- Miocene and Pliocene sediment
- NE : high / SW : low
- Not contain volcanic Rx.
- **KM2** and KM3 : lignite
- NE – SW faults

13.3.2 Geology of the Soma Basin



THICKNESS (m)	LITHOLOGY	DESCRIPTION	GOLDER ASSOCIATES 1983		GÖKÇEN 1982		
			FORMATION		FORM. N	LITH. UNIT	AGE
0-23		Waste dumps	FORMATION				
0-20		Talus					
96-300		Marlstone	P2	PLIOCENE	YATAĞAN	N3-B4	PONTIAN
		Lignite (KP2)					
87-170		Lignite lenses (KP1)	P1				
		Lignite (KM3)					
90-250		Freshwater limestone	YATAĞAN (M3)			N3-B3	PANNONIAN
		Marlstone					
5-140		Marlstone	SEKKÖY (M2)	MIOCENE		N3-B2	SARMATIAN
5-40		Lignite and carbonaceous shale (KM2 member)					
10-40		Conglomerate, sandstone siltstone, and carbonaceous shale	TURGUT (M1)			N3-B1	
		Unconformity					
		Limestone		MESOZOIC			

(fig. 13.4)

Stratigraphy of the basin

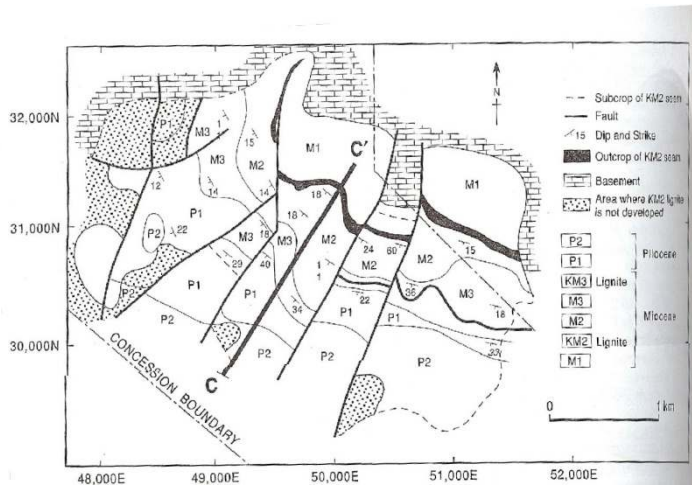
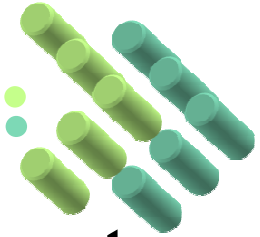
- Top of basement : consist of debris flow
- KM2 : hard, black, bright, cleats, concoidal fracture
- Sharp contact : KM2 – SEKKOY (marlstone*)

Difference Gamma value

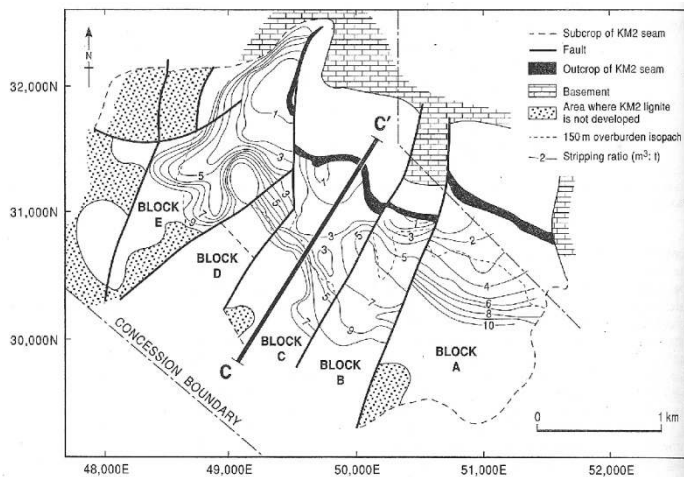
∴ change to an arid climate

*Marlstone : 이회암

13.3.3 Structure of the Soma Basin



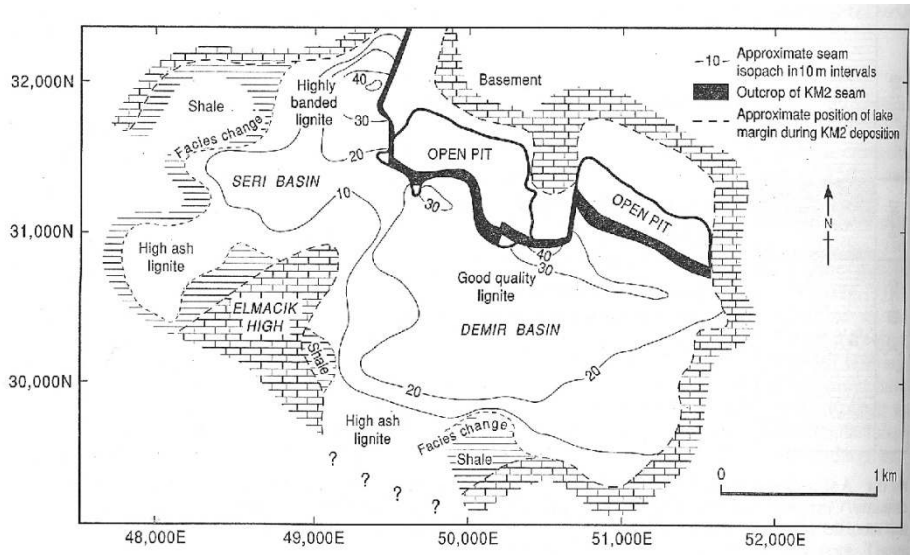
(fig. 13.3)



(fig. 13.8)

- **Tilting to the southwest**
- **Faulting trends NE-SW**
- **Define mining blocks :
A,B,C,D, and E**
- **Dip to the southwest at
an average of 20 degree
north rim → steeper**

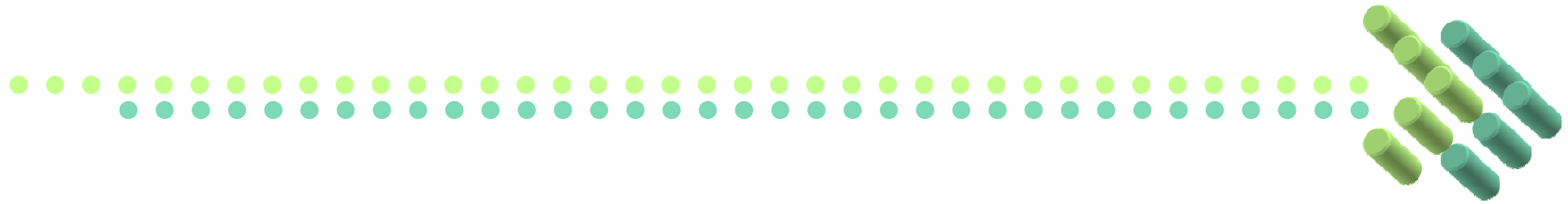
13.3.4 Depositional model for the Soma Basin



(fig. 13.5)

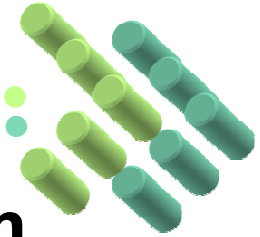
- 북쪽의 가파른 **basin**
- ➔ **run-off** 발생 (모래, 자갈)
- ➔ 중앙으로 유입되어 퇴적
- ➔ **gravelly sandstone at the base to siltstone and mudstone at the top** (fig.13.4)

- ➔ Temporal variation
- ➔ Repetition of fining-upward sequences and rapid lateral variation (filled basin)
- ➔ Plant growth
- ➔ **Form KM2**
- ➔ Climatic change
(warm humid ➔ arid)
- ➔ Sedimentation of **marl** deposition
- ➔ The fine grain size and the calcareous nature of the material suggests deposition by low energy input into a low energy water body
- ➔ **unconformity**



13.4 Data Assessment

13.4.1 Structure contour maps



- **Top of KM2 seam**

- 지하구조의 형상을 알 수 있음

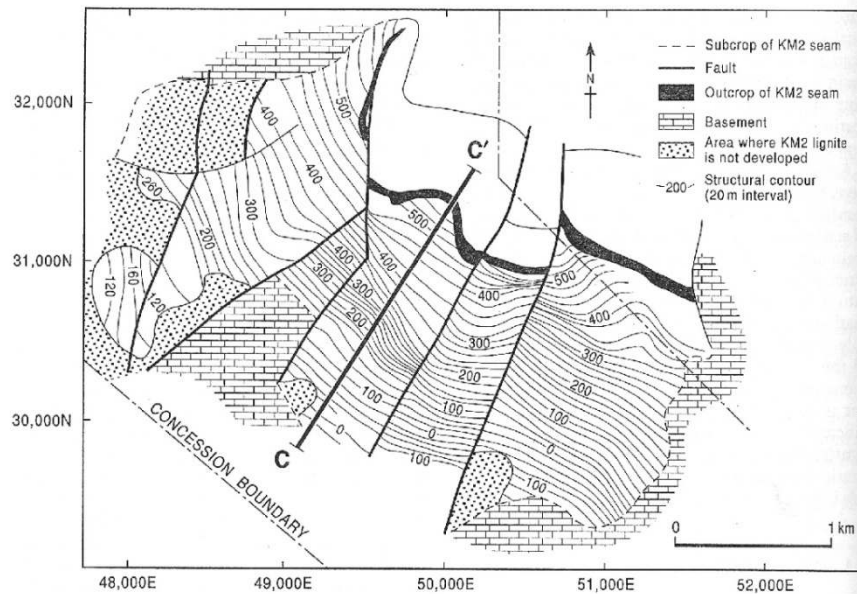
- 단면도와 함께 이용

- ➔ Basin의 단층 패턴 분석

- ➔ 작업의 안정성 도모

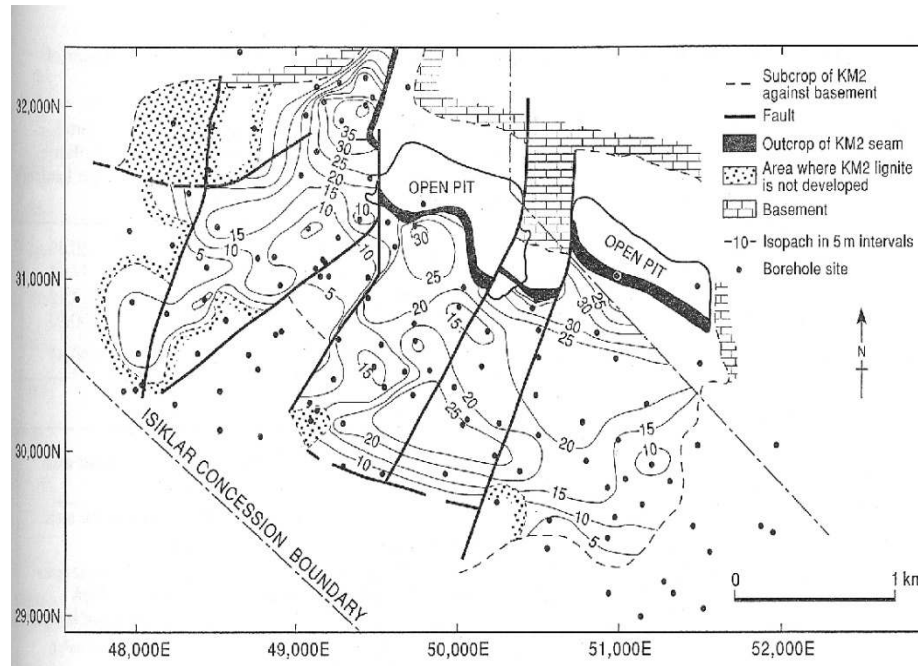
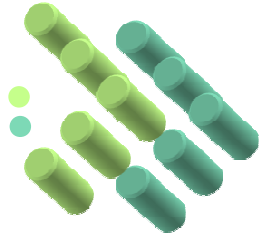
- Mine design 시 이용

- ➔ Identify potential slope failure



(fig. 13.6)

13.4.2 Isopach maps



(fig. 13.7)

I : for the KM2 seam (fig.13.7)

II : for the overburden material

III : for the basement

- **Thickness from 2 to 57 m**

- **Average 17 m**

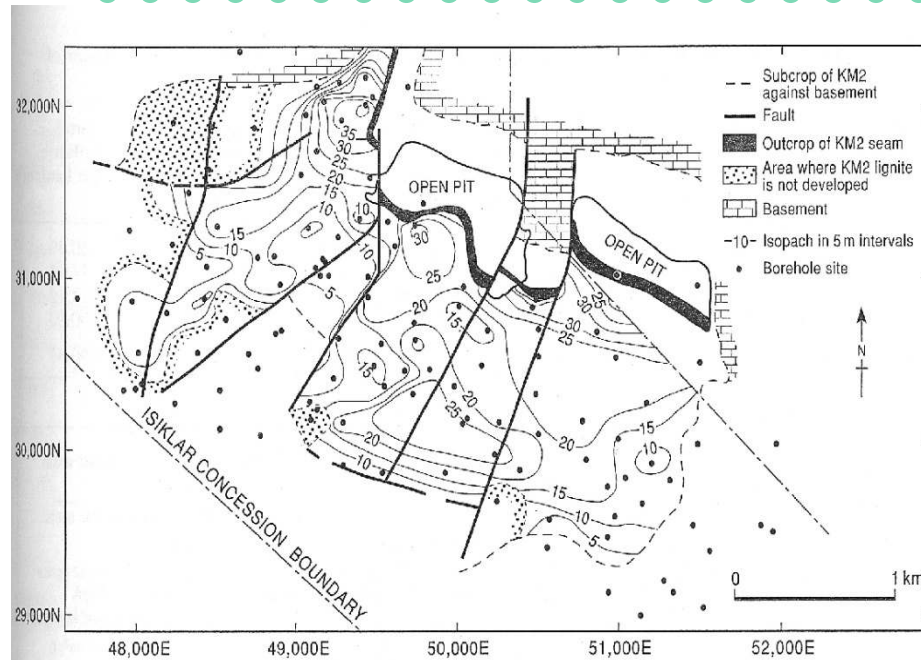
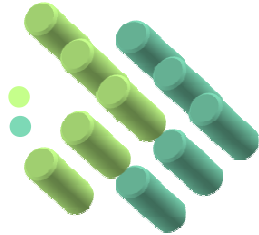
- **(지표의 고도) - (top of KM2)
= (thickness of over burden)**

- **(top of KM2) - (top of base)
= (thickness of KM2)**

- **(top of base) – (bottom of base) = (thickness of base)**

➔ Estimate overburden volume (table 13.2)

13.4.2 Isopach maps

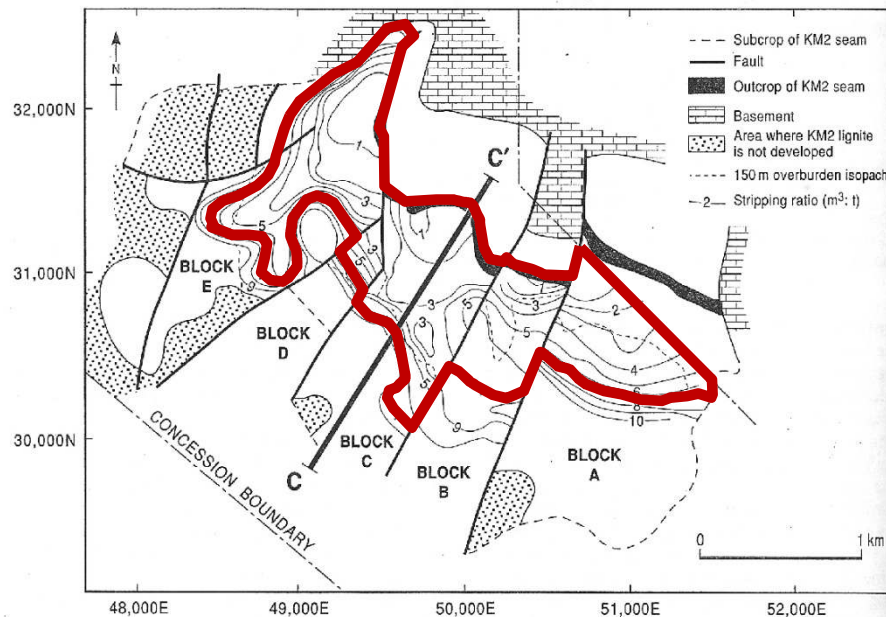
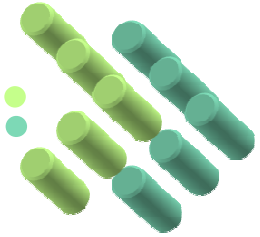


(fig. 13.7)

(table. 13.2)

Block	Pit area			Slopes			Total volume (M bank m ³)
	Area (km ²)	Thickness (m)	Volume (M bank m ³)	Area (km ²)	Thickness (m)	Volume (M bank m ³)	
A	0.146	78.9	11.52	0.148	62.7	9.28	20.80
B	0.412	90.1	37.12	0.189	54.6	10.32	47.44
D	0.072	87.6	6.31	0.096	60.3	5.79	12.10
E	0.882	84.8	74.79	0.357	70.1	25.24	100.03
Totals (mean)	1.512	(85.8)	129.74	0.790	(64.1)	50.63	180.37

13.4.3 Stripping ratio map

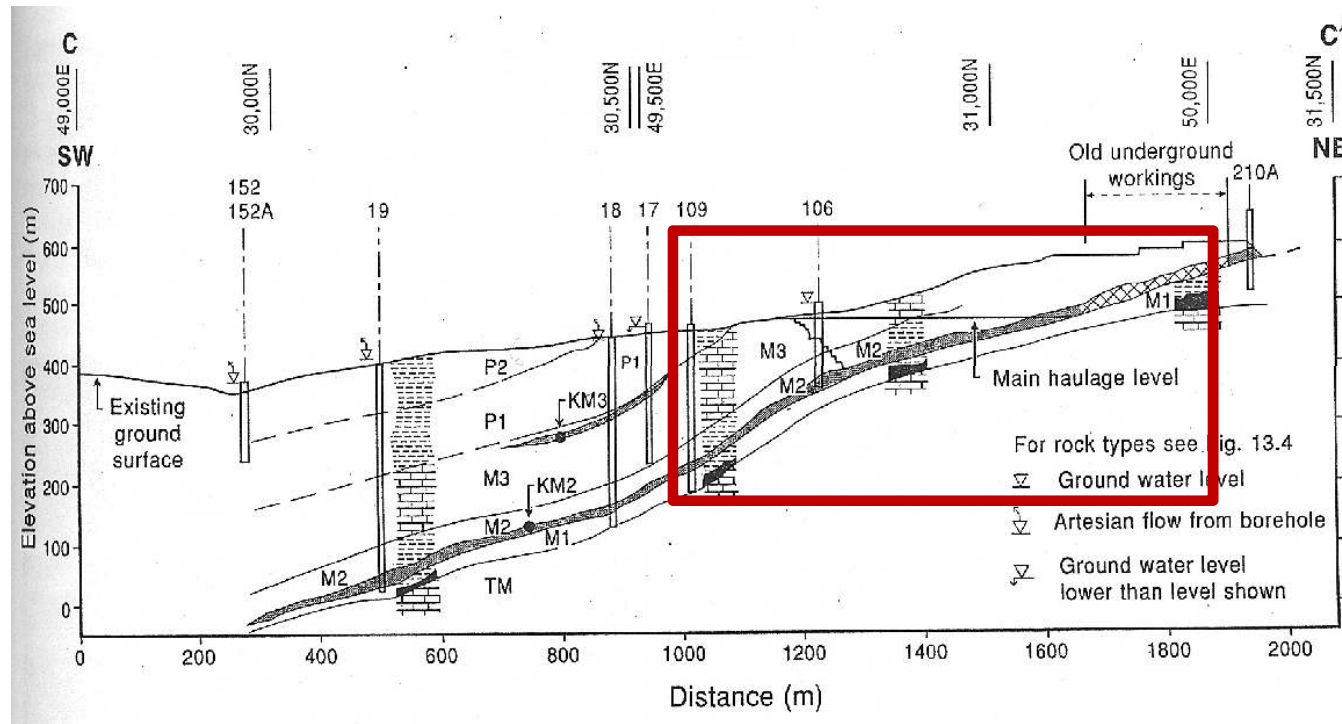
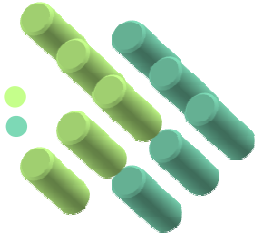


(fig. 13.8)

- **Stripping ratio (or overburden ratio)**
= (overburden material in m³) : (lignite [ton])
= (overburden material in m³) : (lignite (A * thickness)* 1.73)

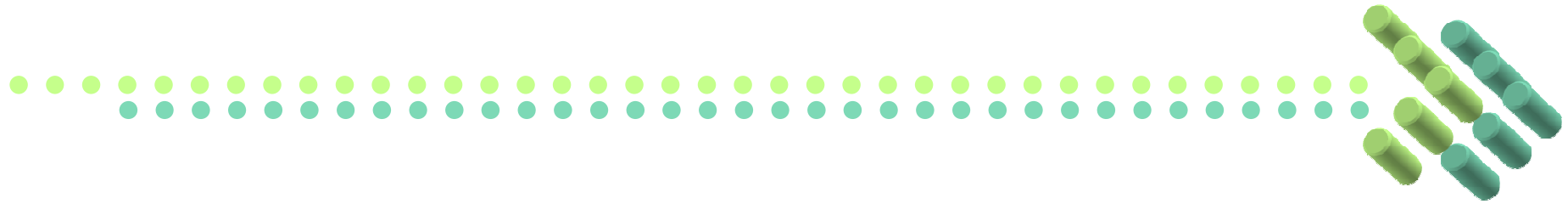
Generally [1 to 10] :1, <7:1 → economically sound

13.4.4 Cross-section



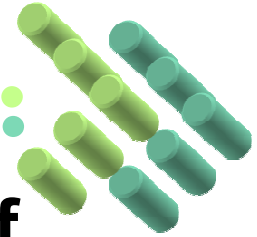
(fig. 13.9)

- To verify the structural information and the seam correlation



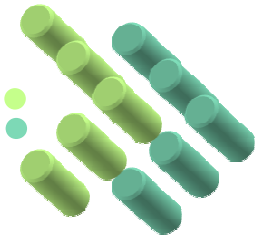
13.5 Geotechnical Investigation

13.5.1 Investigation program



- **To avoid expensive duplicate drilling of boreholes**
 - 1. Geotechnical logging**
→ RQD, point load strength, UCS
 - 2. Geotechnical testing**
 - 3. Structural mapping**
 - 4. Measurement of water levels**
 - 5. Sensitivity analyses**
(e.g. 지하수 조건, 암강도, 전단강도)

13.5.2 Geotechnical conditions



- **Overburden**

- Limestone and marlstone : strong, USC=80 Mpa
mudstone and siltstone : weak, USC < 30 Mpa
- Bedded and jointed, parallel to lignite
- Dips range between 15 and 45 degrees to the SW

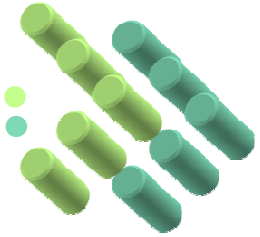
- **Lignite**

- Point load strength = 1 MPa
But, cleated and friable → < 1 MPa

- **Footwall** (highly depends on groundwater conditions)

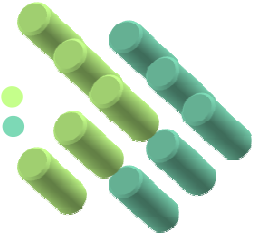
- Mudstone, siltstone, sandstone overlying limestone
Shear strength = 0-30 kPa
angle of shear resistance = 14-17 degree

13.5.3 Hydro-geological conditions

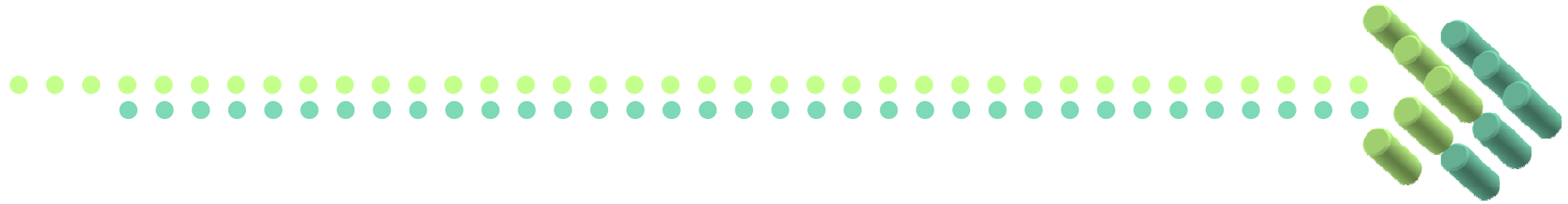


- **By using piezometer**
- **Water level : close to the surface**
- **In lower borehole : occurs artesian flow**
(7 L/sec)
- **Flow maintained for long periods**

13.5.4 Implications for open pit mining

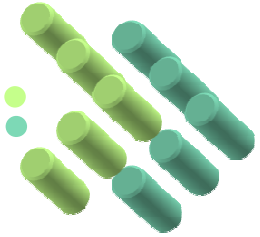


- **Overburden rocks are strong**
→ **Drilling and blasting prior to excavation**
- **최대사면경사 : 약 55 도, 전체경사 : 약 45도**
Haul road 폭 : 약 20 m,
advancing face의 최대각 : 40 도 (fig.13.15)
- **Dewatering would be required**



13.6 Lignite quality

13.6 Lignite quality

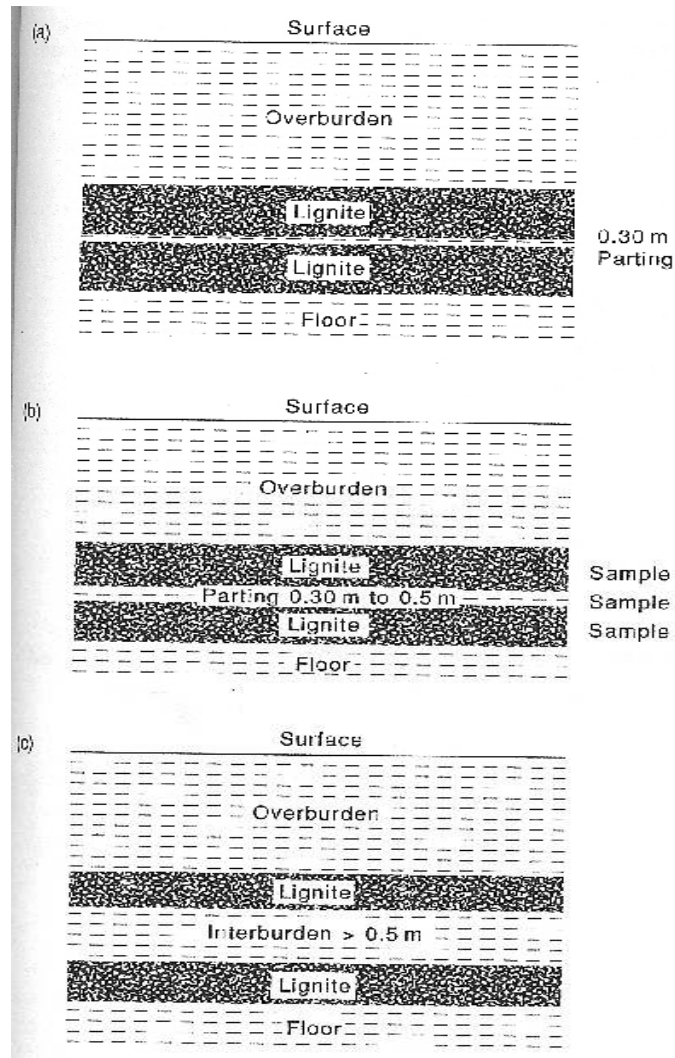
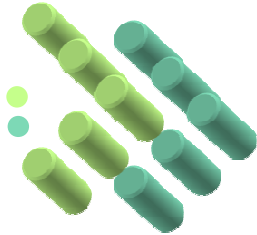


- **Multiple layer, interbedded sediments**
→ **difficulties when seam quality is considered**
- **Alternatives**
 - Nonselective mining : zone within lignite layer is mined in total **without any attempt at selectively mining waste parting**
 - Selective mining : to consider mining the **lignite layers selectively**, aiming to produce a run-of-mine product which is of acceptable quality.

→ **Selective mining**

∴ **lignite quality is low, further quality losses caused by bulk mining would be unacceptable**

13.6 Lignite quality



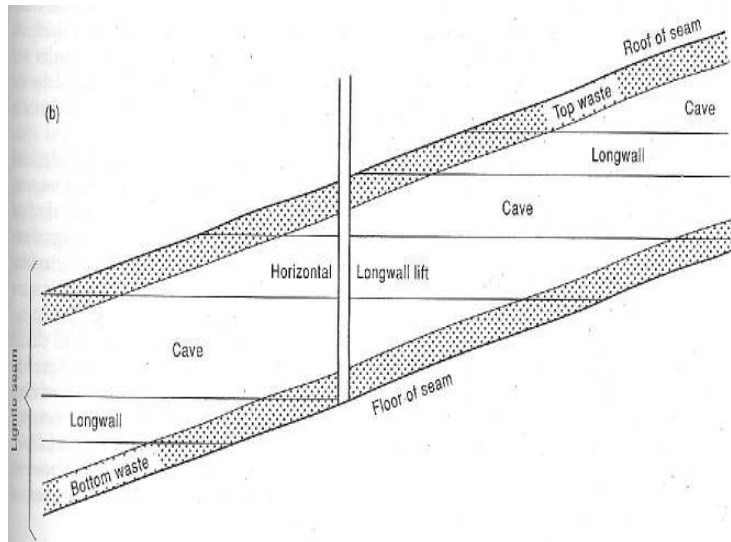
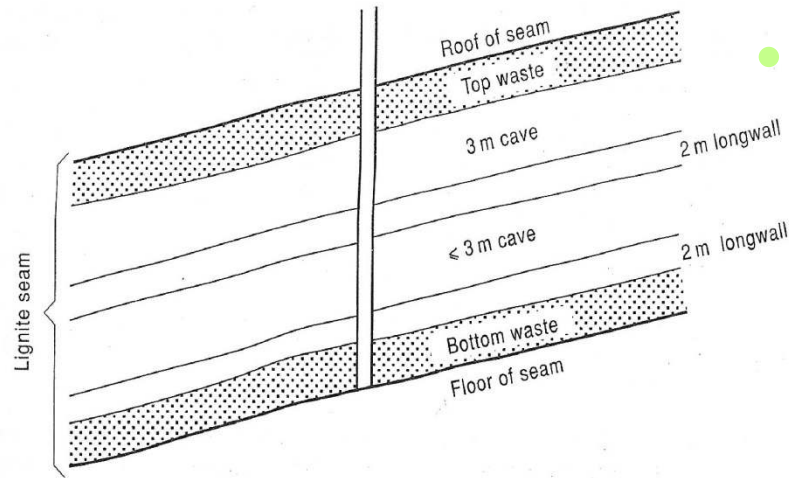
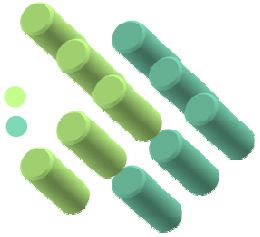
(fig. 13.11)

- **Selection criteria – surface mining option**

Except :

- **Waste parting > 50 cm**
- **Lignite pile < 30 cm with partings on either side**
→ **(total lignite thickness)+ (partings above and below) > 50 cm**

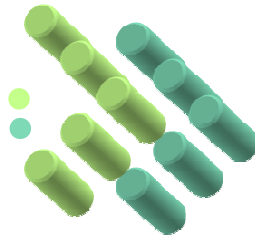
13.6 Lignite quality



(fig. 13.12)

- **Selection criteria – underground mining option**
 - **In-seam mining**
Rejected : (parting) > 1.5m thin lignite & thick waste
 - **Cross-seam mining**
Less selective

13.6 Lignite quality



(table. 13.4)

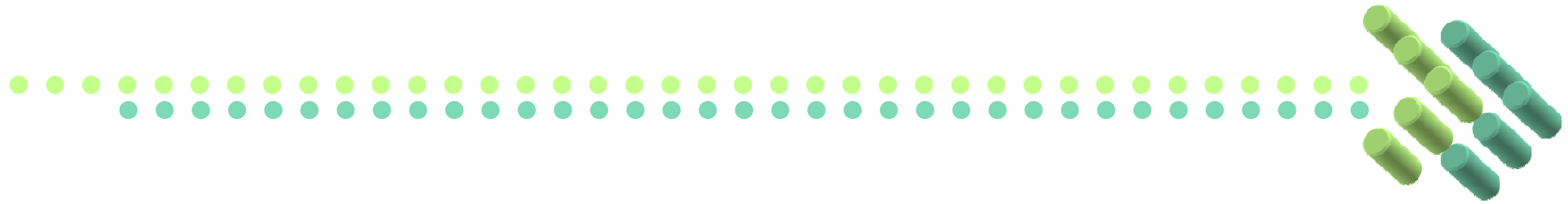
	Borehole number	
	210	320
Lignite, vertical thickness (m)	18.40	11.80
Waste rejected from seam	0.00	3.00
Mineable vertical thickness (m)	18.40	8.80
Number of waste partings	0	3
Number of interface	2	8
Dilution (m)	0.2	0.8
<i>In situ</i> CV (kcal kg ⁻¹)	3524	2071
Undiluted CV (kcal kg ⁻¹)	3524	2679
Mineable CV (kcal kg ⁻¹)	3485	2435
<i>In situ</i> ash content (%)	27.6	39.9
Undiluted ash content (%)	27.6	29.7
Mineable ash content (%)	28.1	33.8
<i>In situ</i> SG	1.55	1.90
Undiluted SG	1.55	1.70
Mineable SG	1.56	1.78
Recovery by thickness (%)	100	75

(table. 13.5)

	Borehole number	
	218	302
Lignite vertical thickness (m)	14.70	12.10
Lignite in waste partings (m)	0.00	0.75
Waste rejected from seam (m)	0.00	2.35
Mineable vertical thickness (m)	14.70	9.00
<i>In situ</i> CV (kcal kg ⁻¹)	3662	2400
Mineable CV (kcal kg ⁻¹)	3595	2500
<i>In situ</i> ash content (%)	24.5	39.5
Mineable ash content (%)	25.2	37.6
<i>In situ</i> SG	1.49	1.73
Mineable SG	1.51	1.70
Recovery by thickness (%)	100	74

(table. 13.6)

	SG	CV (kcal kg ⁻¹)	Ash (%)	Moisture (%)
Open pit				
<i>In situ</i>	1.70	3069	32.7	13.5
Mineable	1.73	3103	33.0	13.5
Block A	1.73	4143	20.9	13.8
Block C	1.73	3104	32.5	14.4
Block D	1.73	2516	39.1	14.0
Block E	1.73	2816	36.6	12.9
Underground				
<i>In situ</i>	1.66	3337	30.0	16.2
Mineable	1.66	3467	27.5	16.2



13.7 Lignite reserve estimates

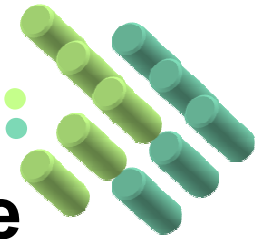
13.7.1 Comparison of estimation methods



Method	Tonnage
Polygons	117.8 Mt
Manual contouring (linear itp.)	102.5 Mt
Kriging (150*150 m block)	109.4 Mt
Statistical mean (area * average thickness)	117.5 Mt

- **Polygon : quick, but overestimated → increase sample density**
- **Manual contouring**
- **Kriging : the best estimation w/ the lowest estimation variance → knows how reliable each block estimate is.**

13.7.2 Confidence in the reserve



- **The confidence in the mineable reserve estimate for the open pit area :**

$$tS/n$$

S : standard deviation, **n** : number of borehole

t : t-value for n-1 DoF at the 90% confidence lv.

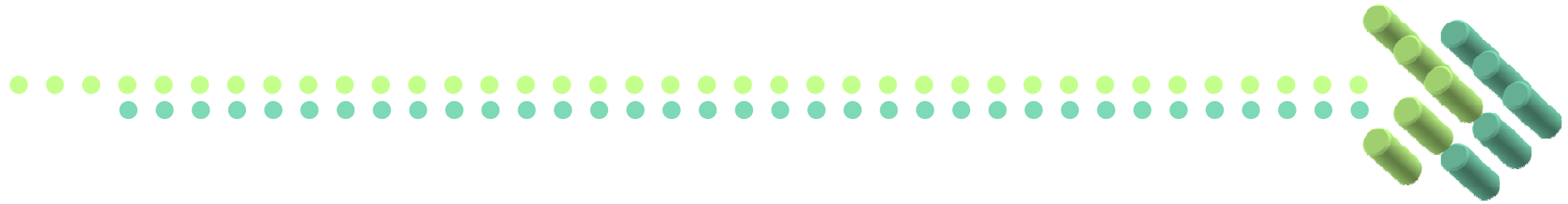
- **Global confidence in the average expressed as a percentage :**

$$G = A^2 + B^2$$

A : % of the mean for thickness

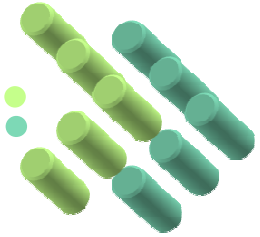
B : % of the mean for SG

$$A=15.9\%, B=5.9\% \rightarrow G=17.0\%$$



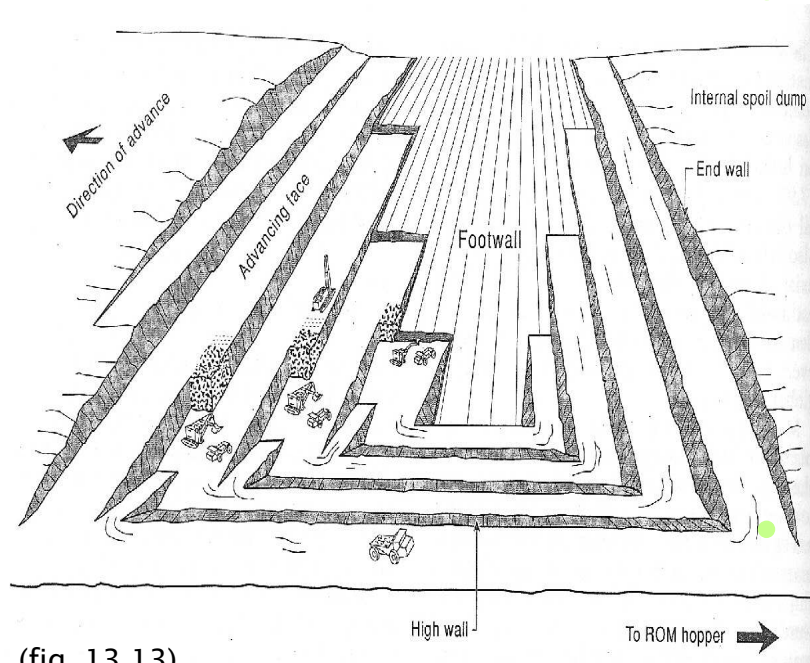
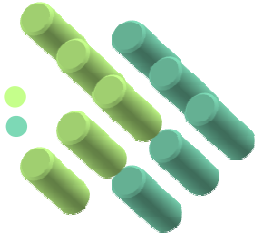
13.8 Surface mine evaluation

13.8 Surface mine evaluation



- **This evaluation project called for a production rate of 2.13 Mt ROM(run-of-mine) lignite each year**
- **After rejection of some waste dilution
→ 2 Mt per year**
- **Exploit the mineable reserves over a mine life of 21 years.**
- **Block B is not advisable
∴ Lignite has been partly extracted**

13.8.1 Selection of mining method



(fig. 13.13)

- Three alternative

1. Advance down the dip
 2. Advance up the dip
 3. Advance along the strike (terrace mining)
- 3rd is the suitable

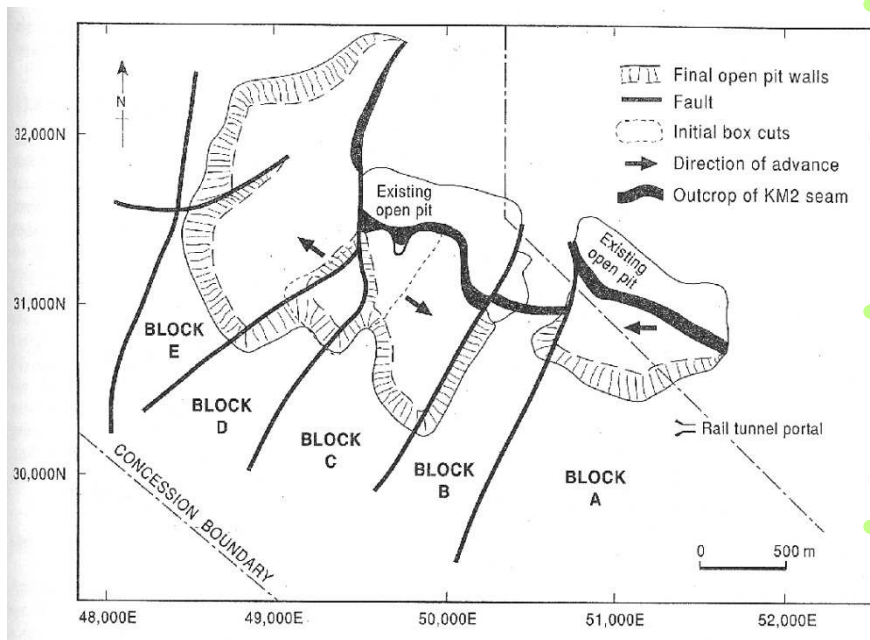
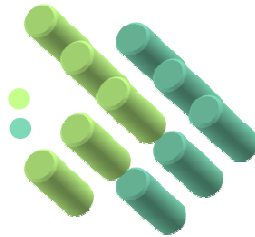
Until

- Economic stripping limit
- (Practical) mining depth limit

- Advantages of 3rd method

1. Minimum area of footwall clay will be uncovered at any time
→ reducing the risk of footwall failure
2. Internal dumping of waste reduces transport costs, helps stabilize the footwall, and begins reclamation at an early stage of mining
3. The stripping ratio is constant over the life of the mine → mining costs stabilized

13.8.2 Mine design



(fig. 13.14)

Box-cut locations

- **First : block C & D**
- **Close to center of the reserve**
- **Second : block A**

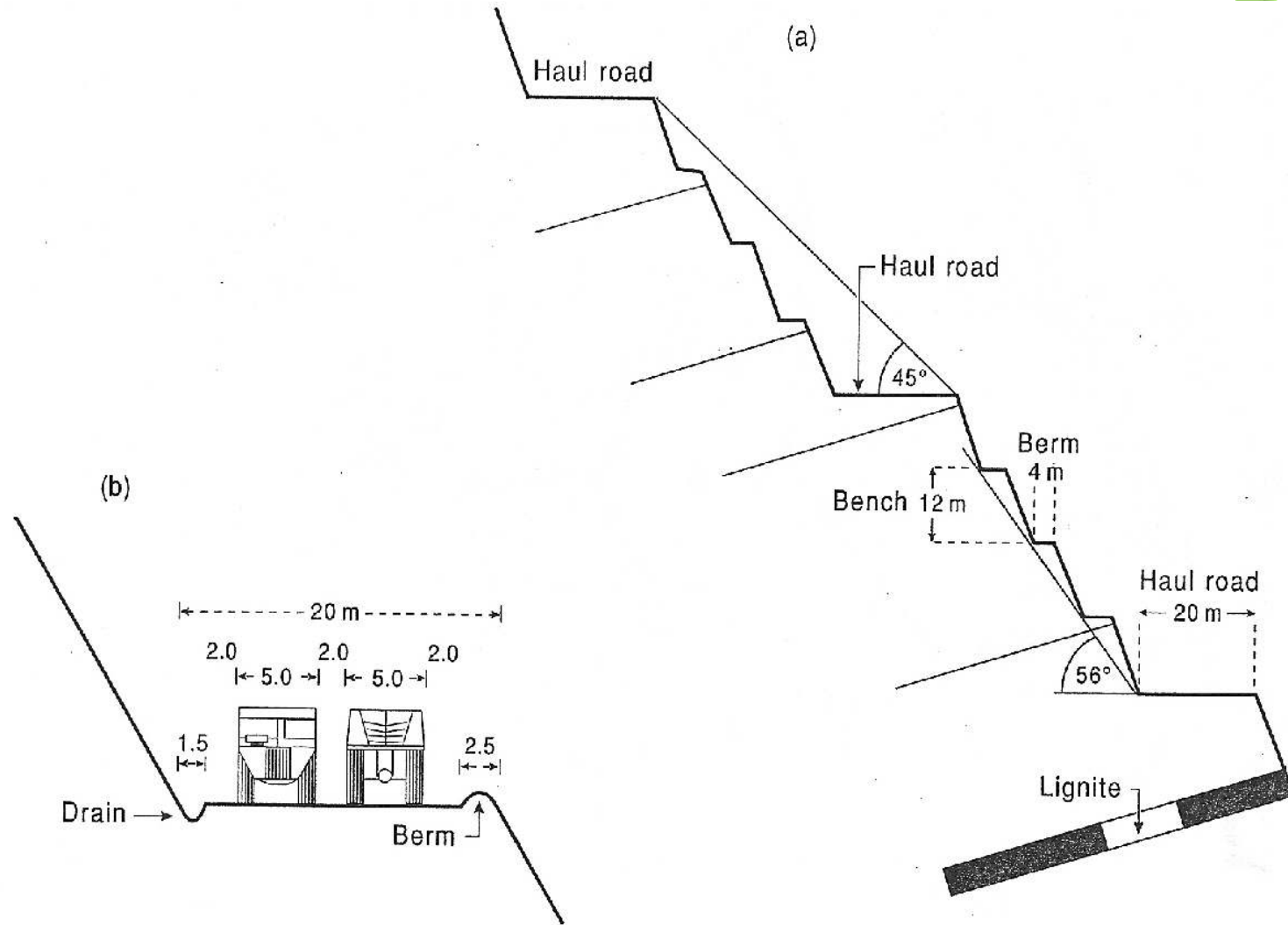
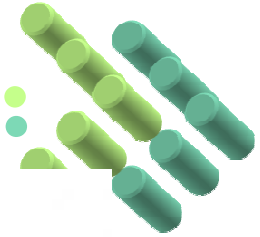
Slopes and access

- **Bench height = 12 m**
(fig. 13.15)

Haul roads

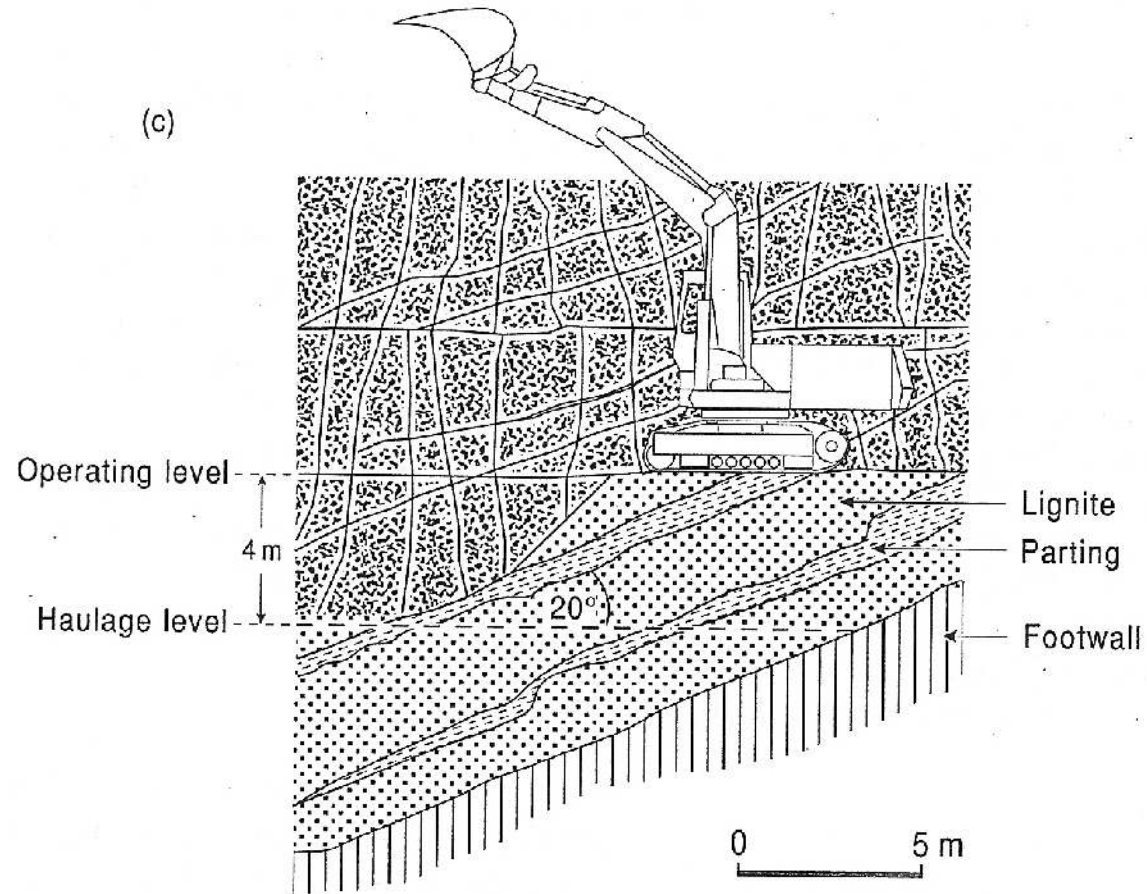
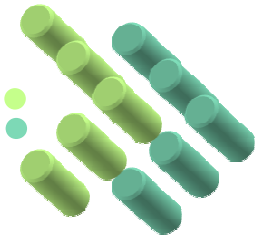
- **Width = 20 m**
- **Spoil disposal**

13.8.2 Mine design



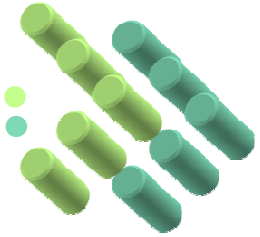
(fig. 13.15)

13.8.2 Mine design

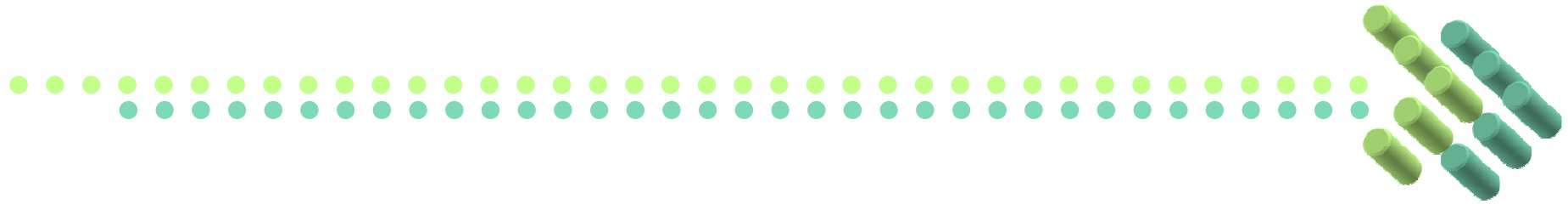


(fig. 13.15)

13.8.3 Mining equipment

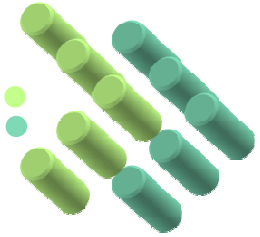


- **Ratary blasthole rigs (250-mm-diameter holes)**
- **Electrically powered rope shovels
: overburden stripping**
- **Hydraulic face shovel : lignite loading**
- **Rear dump trucks : transport**
- **Bulldozer**



13.9 Summary

13.9 Summary



- **Available data on the reserve tonnage and quality of the Soma deposit were adequate for a feasibility evaluation.**
 - ➔ **would support a projected life of 21 years**
- **Additional data were required on the groundwater regime before forecasts**
 - ➔ **Effects of groundwater pressure and flows on mining**

THANK YOU

