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CENTRAL COALFIELDS LIMITED (CCL)

MINING PLAN AND MINE CLOSURE PLAN OF SELECTED DHORI (LOWER) OPEN CAST MINE (PROJECT AREA-211.82 HA, RATED CAPACITY-2.0MTY)

EAST BOKARO COALFIELD

DIST – BOKARO (JHARKHAND)

[TEXT & PLANS]

NOV 2021

REGIONAL INSTITUTE-III

CENTRAL MINE PLANNING & DESIGN INSTITUTE LIMITED (A Subsidiary of Coal India Limited) GONDWANA PLACE, KANKE ROAD, RANCHI – 834 008, JHARKHAND, INDIA

INDEX OF CHAPTERS OF THE MINING PLAN (INCLUDING MINE CLOSURE PLAN)

SI No.	Chapters	Page No
	Summarized Data	4-7
Chapter 1	Project Information	8-11
Chapter 2	Exploration, Geology, Seam Sequence, Coal Quality and Reserve	12-19
Chapter 3	Mining	20-26
Chapter 4	Safety Management	27-36
Chapter 5	Infrastructure Facilities proposed and their Location	37-38
Chapter 6	Land Requirement	39-40
Chapter 7	Environment Management	41
Chapter 8	Progressive & Final Mine Closure Plan	42-44

LIST OF PLATES

SI. No.	Particulars	Scale				
MINING	MINING					
1.	Final Stage Quarry Plan	1: 4000				
2.	Quarry Cross-Section along A-A', B-B', C-C'.	1: 2000				
3.	Final Stage Dump Plan	1: 4000				
4.	Key Land use Plan	1: 4000				
5.	Post Mining Land Use Plan	1: 4000				

SUMMARISED DATA

SI. No.	Particulars	Unit	Value
Α.	GENERAL		
1	Name of Project		Selected Dhori(lower) OCM
2	Name of Area / Company		Central Coalfield Ltd.
з	Nearest Railway Station from project	Name	Phusro Railway Station
5	Nearest Naiway Station nom project	km	2
Λ	Nearest National / State Highway /	Name	NH 2 / Dumri-Bermo-Jaina road
4	Approach Road	km	35 / 4

В.	GEOLOGICAL		
1	Name of geological blocks considered	Name	Selected Dhori
2	Area of the geological blocks	sq. km	3.00
3	Borehole Density within blocks	BHs / sq.km	9
4	Description of all coal seams within block		

Stratigraphic Sequence	Thickness (m)	Net Geological Reserves (MT)	Remarks
Comb. Karo Gr. of seams Karo VI to X	57.33-69.28	85.17	Under exploitation, not considered.
Karo V	2.92-4.93	7.01	Reserve not assessed in Geological Report
Karo IV	0.18-1.15	-	Reserve not assessed due to low thickness.
Karo III	3.25-7.66	12.75	
Karo II	0.17-1.25		Not opencastable due to low thickness.
Karo I	0.30-0.40		Not opencastable due to low thickness.
TOTAL (Geological reserve considered for mining)		19.76	

Note: Reserves of Seams Karo V & III has only been considered.

SI. No.	Particulars	Unit	Value
C.	TECHNICAL		
1	Area of the proposed mine block (only the quarry area)	sq. km	1.79
2	Borehole density within mine area	BHs/sq. km	12.5
3	Mine parameters Extent along strike (min. – max.) Extent along dip (min max.)	m	600 – 1300 800 – 1450
4	Description of coal seams proposed to be worked along with the parting details		

Name of seam/ Parting	Thickness range considered (m)	Av. Thickness / Parting Thickness (m)	Avg Grade	Mineable Reserve (MT)	Volume of OB (Mcum)
Rehandling OB	3.00 - 48.50	12.20			12.34
Seam VI to X combined	57.33-69.28	-	W III/IV	_*	
Parting V - VI to X	10.20 - 30.25	17.40			32.85
V	0.15 - 5.04	2.00	G 9	4.51	
Parting IV - V	5.34 – 14.15				20.81
IV	0.10 - 1.30	17.80			
Parting IV - III	4.40 - 5.10				
	1.50 - 7.22	3.50	W III	6.21	
Total Coal OB				10.72	66.00

* Karo VI to X combined has been considered for mining. This seam forms the topmost seam in the proposed mining area. The seam has been mined extensively with an occurrence of fire in the area. Presence of this seam is possible in patches in the proposed mining area. Since its extent and quantity could not be ascertained due to past mining, balance reserve of this seam has not been assessed. However, the same needs to mined out along with the underlying seams Karo V & III.

SI. No.	Particulars	Unit	Value
5	Av. Stripping Ratio	m³/te	6.16
6	Method of Mining		By Shovel Dumper Combination
7	Target Output	MTY	2.00
	Production capacity (at 125%)	MTY	NA
8	Year of achieving Target Production	Year	4 th
9	Year of start of Internal Dumping	Year	1 st

10	Production Phasing (date upto target year)			
Year	1	2	3	4
Coal (MT)	0.50	1.00	1.50	2.00
OB (Mcum)	3.70	7.41	10.18	12.36

11	Total Mine Life	Years	8
	Production build-up period	Years	3
	Production period	Years	3
	Tapering / mine closure period	Years	2
12	Major HEMM Deployed (Max)	Unit	No.
	Diesel Hydraulic Shovel	10-12 Cum	2
	Diesel Hydraulic Shovel	5.5-6.5 Cum	2
	Diesel Hydraulic Backhoe	4.3-5.0 Cum	1

Rear Dumper	100 T	12
Rear Dumper	60 T	14
Diesel RBH Drill	250 mm	1
Diesel RBH Drill	160 mm	3
Dozer	410 HP	3
Dozer with Ripper Attachment	410 HP	1

SI. No.	Particulars	Unit	Value
13	Seam-wise weighted average grade of coal (non-coking/coking)	Non-Coking /Coking	Grade
	Seam V	Non-Coking	G 9
	Seam III	Coking	W III

SI. No.	Particulars	Unit	Value
1/	Presence of Major Surface Constraints	Nalas & Nadi	Tisri Nala, Joria Nala, public road &
14	(nallas, road, power line, etc.)	Indias & Indui	existing mine workings.
	Coal Transport within the mine		
15	(In-pit belt conveying system or by		Dumper
	Truck)		
	Surface Coal Transport to		By conveyors from CHP to bunkers,
16	Siding/Despatch Point and Mode of		mechanized loading through
	Despatch		Hopper at Tarmi siding.
			Tarmi railway siding at Bhandaridah
17	Any Railway Siding and distance		Railway station is about 1-2 km
			away
18	Name of any Specific Customer/Industry		Basket linkage

D. ENVIRONMENTAL & OTHERS

SI. No.	Particulars	Unit	Value
	Total Land required		211.82
	Land acquired		211.82
1	Total Land to be acquired	Ha	Nil
	Non-Forest land		Nil
	Forest land		Nil
	Land to be acquired within minetake area		
2	(excavation area)	L la	Nil
	Non-Forest land	на	
	Forest land		

SI. No.	Particulars	Unit	Value
3	Land to be acquired outside minetake area (Beyond Excavation Area, such as Approach Road, Infrastructure, Colony, nala diverson etc.)	На	0.00
4	Land to be acquired for external dumping	На	0.00
5	Habitation & Rehabilitation No. of villages within mine boundary No. of PAFs to be rehabilitated	Nos.	0 0
6	Drainage of the Area (Name of river/nala)		North-south flowing Tisri Nala & Joria Nala discharging to West to east flowing Damodar River.

CHAPTER 1: PROJECT INFORMATION

	Parameters	Details
1.1	INTRODUCTION	
1.1.1	Name of Coal Block	Selected Dhori
1.1.2	Name of the Coalfield	East Bokaro Coalfield
1.1.3	Base date of Mining Plan/ Mine Closure Plan	Nov 2021
1.1.4	Linked End Use Plant	Basket
1.1.5	Distance of End use plant from the pit head of the project in "km"	1.0 km
1.1.6	Mode of Coal Transport	Road/ Railway

1.2 LOCATION, TOPOGRAPHY AND & COMMUNICATION

1.2.1	Location of coal deposit (District and State)	District - Bokaro , State - Jharkhand
1.2.2	Communication: PWD roads, railway lines, Air	Selected Dhori OCM is well connected by rail and road. The Gomoh- Barkakana loop line of the Eastern Railway passes adjacent to the area. The nearest railway Station 'Phusro' on the Barkakana-Gomoh loop line is about 2 km to the west of the area. The area is well connected by an all-weather Phusro-Dumri metalled road up to the existing Tarmi opencast project of CCL. The block is about 30km from Bokaro, about 80km from Dhanbad & about 110km from Ranchi. The nearest commercial airport is situated at Ranchi.
1.2.3	Availability of power supply	It is envisaged that this project will receive power at 33 kV from Bokaro Thermal Power Station (BTPS) of DVC. For this, two nos. existing 33 kV overhead lines from BTPS to B&K Main-sub-station shall be extended up-to proposed quarry.
1.2.4	Prominent physiographic features, drainage pattern, natural water courses, rainfall data, highest flood level	The area witnesses a sub-tropical climate. The average temperature varies from 43° Celsius to 49° Celsius during summer and 4° Celsius to 10° Celsius during winter. Heavy rainfall occurs in the month of June to September. The average rainfall varies from 1200-1400mm of rainfall / annually. The area is hilly with isolated hillocks and intervening plain lands and valleys occupied by paddy fields, quarters and dumps of some existing and abandoned quarries. The river Damodar serves as the perennial source of water for the area. The Tisri Nala, a prominent tributary of Damodar, flows from north to south to the west of the area.
1.2.5	Important surface features within the project area and	North-south flowing Tisri Nala & Joria Nala discharging to West to east flowing Damodar River.

Parameters	Details
major diversion or shifting involved	

1.3 DETAILS OF THE ALLOTTMENT AGREEMENT

1.3.1	Name the Allottee	-NA-
1.3.2	Details of allotment/vesting order	-
1.3.3	Name and address of the applicant	-
1.3.4	Name of the Previous allottee of the Block	-
1.3.5	Starting Date of the Mine as per CMDPA	-
1.3.6	Rated Capacity as per CMDPA	-
1.3.7	Production Schedule as per opening permission (meeting provisions of CMDPA if any)	-
1.3.8	End Use of Coal as per allotment order if any	-
1.3.9	Cardinal Points co-ordinates of the Block boundary	_

1.4 DETAILS OF THE PREVIOUS APPROVAL OF MINING PLAN

1.4.1	Date of Approval	04.11.2019				
1.4.2	Conditions, if any	-				
1.4.3	Scheduled year of start of production	1 st Year				
1.4.4	Proposed year of achieving the targeted production	5 th Year				
1.4.5	Date of actual commencement of mining operations, if operations already started	-				
1.4.6	Likely date of mining operations, if operations not yet started & reasons for non- commencement of operations	-				
1.4.7	Planned production and actual levels achieved in last 3		Year wis	se Production	n Details	
	vears (Coal in Mte, OB in				OB MM ³	
	MM ³ , SR in MM ³ /te)	Year	Coal "Mte"	Solid	RH	Total OB
		2018-19	3.504	0.681	0.085	0.766
		2019-20	2.216	-	1.502	1.502
		2020-21	1.590	-	2.026	2.026

	Parameters	Details
1.4.8	Reasons for difference between the planned and actual production levels	_

1.5 PARAMETERS OF APPROVED MINING PLAN VIS-À-VIS PROPOSED MINING PLAN

		Approved Mining Plan	Proposed Mining Plan
1.5.1	Block Area in "Ha"	211.59	179.44
1.5.2	Block Area Projectised "Ha"	211.59	179.44
1.5.3	Lease area "Ha"	-	-
1.5.4	Project Area "Ha"	264.85	211.82
1.5.5	Life of the Project "Yrs"	11	8
1.5.6	Minimum and Maximum Depth of working "m"	25 m-146 m	10 m-120 m
1.5.7	Net Geological Block "Ha"	300	300
1.5.8	Production Target "MTPA"	2.0	2.0
1.5.9		Top to bottom	Top to bottom
	Seams Available "As per GR"	 Karo VI to X combined Karo V Karo IV Karo III Karo II Karo I 	 Karo VI to X combined Karo V Karo IV Karo III Karo II Karo I
1.5.10	Seams not considered for Mining with Reasons	Seam IV, Seam II and Seam I. The very low thickness. Karo VI to X combined has been con the topmost seam in the proposed mined extensively with an occurrent seam is possible in patches in the pr and quantity could not be ascertaine not been assessed. However, the s	ese seams are not mineable due to nsidered for mining. This seam forms d mining area. The seam has been ce of fire in the area. Presence of this roposed mining area. Since its extent ed due to past mining, its reserve has same needs to mined along with the
1.5.11	Gross Geological Reserve		
4 5 40	"Mt"	07.04	40.70 M
1.5.12		27.81	
1.5.13		-	9.04 Mt
1.5.14	IVIINADIE Keserve "Mt"	16.83 Mt	10.72 Mt
1.5.15	Extractable Reserves "Mt"	16.83 Mt	10.72 Mt
1.5.16	% of Extraction/ recovery	100%	100%
1.5.17	Reserve Depleted (till the base date) Reserves " Mt"	Nil	Nil

	Parameters	Details		
1.5.18	Balance Extractable reserve "Mt"	16.83	10.72 Mt	
1.5.19	Average Grade	Grade – W-III' for Karo Seam	-III and 'G9' for Karo Seam-V	
1.5.20	OB in MM3	117.28	66.00	
1.5.21	SR MM3/te	6.97	6.16	
1.5.22	Mining Technology	Shovel-dumper mining system has t	been envisaged for working this OC	
		mir	ne.	
15.22	Coal Panaficiation onvisaged	Part anal (asking asa	I) to poorby Washany	
1.5.25		Fait coal (coking coa		
1.5.24	Handling of Rejects			
1.J.2J	Eand use pattern Tha	211 59	179 44	
1.	External Dump	211.00	-	
2.	External Dump	-	-	
3.	Safety Zone/Green Belt	29.76	24.15	
4.	Other Use (Nala, embankment)	-	3.17	
5.	Infrastructure area (CHP, Haul Road, Colony)	23.50	5.06	
	Total	264.85	211.82	
1.5.26	Reasons for revision			
As per A	As per Approved Mining Plan of Selected Dhori OCP (for extraction of lower Seams) (02 MTPA/264.85Ha) some			
part of F	part of Project Area comes under GM-JJ land. The GM-JJ land has not been acquired till date. Therefore, this			
mining plan has been prepared excluding un-diverted GM-JJ land. So mining plan of Selected Dhori (for extraction				
of lower	of lower Seams) has been revised (02 MTPA/211.82 Ha) for approval from competent authority.			

CHAPTER 2: EXPLORATION, GEOLOGY, SEAM SEQUENCE, COAL QUALITY AND RESERVE

	Parameters	Details	
2.1	DETAILS OF THE BLOCK		
2.1.1	Particulars of adjacent blocks: North, South, East, West	North: N.A South: Pichri Block	East: Tarmi Block West: Tisri Block
2.1.2	Location of the Block District / State	District- Bokaro State- Jhark	hand
2.1.3	Area of the Block "Ha"		300
2.1.4	Area of the geological block projectized "in Ha" (Area of the geological block considered for liquidation of coal reserve)	2'	11.82
2.1.5	Balance area yet to be projectized "Ha"		-
2.1.6	Likely Reserve in the area yet to be projectized "Mte"		-
2.1.7	Cardinal Point Co-ordinates of the non- coal/lignite bearing area <u>/existing</u> <u>mining lease</u> outside the allotted Geological Coal/Lignite block (Duly certified in line with para 1.9 of the Guideline, if fresh inning lease required)	This project is bounded by the I longitudes 86º02'E to 85º03'E a sheet No. 73 I/1 (RF 1:50,000).	atitudes 23º45'N to 23º48'N and and lies in the Survey of India Topo
2.1.8	Accredited Mining Plan preparing agency (MPPA)if the project area is confined within the vested/allotted block boundary/existing mining lease and Where the project area extends beyond the block boundary, a certificate of Qualified person/ Accredited Mining Plan preparing agency (MPPA)should be supported with a certificate of State Government mines and Geology department must be attached, which should specify (a) intent of the state government for grant of lease beyond the vested geological boundary; (b) non-existence of Coal/ Lignite in the area beyond the vested/allotted geological block boundary/existing mining lease to rule out the issue of encroachment and use of coal bearing area (beyond the vested/allotted block boundary/existing mining lease) in the mining plan		-

	Parameters	Details
	The Project area, Lease area and geological block area in "Ha" shall also be envisaged.	
2.1.9	KML file of the Proposed lease area, Project Area and geological block.	Enclosed as Annexure-
2.1.10	Whether the proposed project area is confined within the allotted block boundary/existing mining lease, if not, the reason for deviation from allotted block boundary, may be given.	Yes
2.1.11	If the project area extends outside the allotted block boundary/ <u>existing</u> <u>mining lease</u> , confirmation about non- occurrence of coal/lignite in the area under reference needs to be furnished	-
2.1.12	Type of the Project (Operating / under Implementation) and year of Starting.	Under operation since nationalization
2.2	EXPLORATION, GEOLOGY AND ASS	ESSMENT OF RESERVE
2.2.1	local geology, structure, stratigraphic sequence, characteristics of the litho- logical units (coal seams /partings/overburden).	The block structurally forms a part of the northern limb of the east west elongated synclinal sub basin of the East Bokaro Coalfield. The general strike of the strata is east west dipping southerly at angles varying from 7° to 12°. However, the strike of the beds changes from NW-SE in the east to NE-SW in the west through E- W in the central part. The Gondwanas are represented by rocks of Talchir, Karharbari and Barakar Formations. Post Barakar formations are absent in the block.
		Talchir Formation: The Talchir Formation is developed in the northern part of the block and occurs as narrow strip. It comprises of pebbly sandstone, conglomerate and fine grained sandstones with laminated greenish shale.
		Karharbari Formation: The Karharbari Formation lies conformably above the Talchirs. Exposures of the Karharbaris are found in the northern part. It is represented by coarse to very coarse grained sandstone with gritty layers and thin coal seams (Karo-I to IV). Among the thin seams found in the Karharbari Formation, the Karo Special Seam Karo-III is most important from quality point of view.
		Barakar Formation: It occupies a major part of the area. It is represented by medium to coarse grained sandstone, grey shales and thick coal seams (Karo-V– X). The coal seam however is highly interbaded.

	Parameters	Details
		Igneous intrusion – Igneous activity has been recorded in the western and southern part of the block. The intrusives are represented mainly by dolerite dykes and mica-lamprophyre. In boreholes EBC-11 & 21 the seam Karo-III has been completely burnt due to igneous intrusion. A steeply dipping dolerite dyke has been exposed in a ravine east of Tisri nala near borehole EBC-11. The dolerite dyke has been encountered in EBC-11 from a depth of 7.50m to 53.80m.
2.2.2	Geology: The Selected Dhori Geological Block co 45' and 23° 48' N and longitude 86° 02' a Coalfield and lies in the Bokaro distric Selected Dhori Block, East Bokaro Coal	overing an area of 3.00 Sq km (approx.) is bounded by Latitude 23 ^o and 86 ^o 03' E. It is located in the eastern most part of the East Bokaro cts of Jharkhand. Geological Report on exploration carried out in field. District-Giridih, Bihar, May 1977, CCL, Ranchi.
	2.2.2.1 Exploration Status	

In total 18 boreholes have been drilled in different phases from 1975 to 1978. Total meterage of 2059.20 m by various agencies. The borehole density of the Geological Block is as follows:

Block	Area (sq km)	No of Boreholes	Meterage Drilled
Selected Dhori	3.0	CCL (CDD I-10)	Sept.1975- May 1976
		GSI (EBC-7,8,9,11, 17,21,24,30)	1967-78
Overall	3.00	18	2059.20

Additionally, 8 boreholes have been drilled recently by CCL to ascertain the data in-sufficiency mostly in the south-western part. However, the quality estimation has not been done in these boreholes. The borehole data within the selected Dhori geological block for seam-III is very low. Hence, the alignment of fault may change during mining.

2.2.2.2 Deposit Structure

4

The strike of the beds in general is East - West with a southerly dip varying from 7° to 12°.

A number of faults have been encountered within the Geological Block. The blocks within the projectised area with their location, extent and amount of throw are enumerated in the table below:

	Parameters	Details			
Fault No	Location & Extent of Fault	Trend of Fault	Amount & Direction of Throw	Evidence of Fault	
F1	Central part of the block merges with F1 of Tisri and abuts with FS7.	E-W	20m Northerly	Based on displacement of outcrop of seams.	
F2	Southern part of the block near borehole CDD4.	NE-SW	10m, Northerly	Seam III omitted and parting between Seam V & II reduced considerably.	
F3	Southern part of the block	ENE- WSW	10m, Northerly	Lower part of Karo group of seams omitted upto 10m.	
F4	Southern part of the block near EBC-21	NE-SW	10m, North Westerly	Displacement of outcrop of Karo group of seams.	
F6	Southern part of the block near CDD-10	ENE- WSW	10m Northerly	Omission of Karo group of seams upto 10m.	
F7	Northern part of the block near MBT-3	E-W	35m, Northerly	Based on stratum contour	
F8	Northern part of the block near EBC-30	WNW- ESE	25m, Southerly	Based on stratum contour	
F9	North-western part of the block	NE-SW	50m, North Westerly	Based on stratum contour	

2.2.2.3 Sequence of Coal Seams and Parting

The details of sequence of coal seam and parting is given below:

SeamBH IntersectionThickness RangeComb. Karo Gr. of seams Karo VI to X12 60.00 57.33 Parting25.0024.50Karo V11 3.00 2.92 Parting10.00 5.34 Karo IV10 <1.00 0.18 Parting 4.50 4.40 Karo III10 5.50 3.25 Parting 4.00 4.00 4.00	ə (m)			
Seam	Intersection	Av.	Minimum	Maximum
Comb. Karo Gr. of seams Karo VI to X	12	60.00	57.33	69.28
Parting		25.00	24.50	27.10
Karo V	11	3.00	2.92	4.93
Parting		10.00	5.34	14.15
Karo IV	10	<1.00	0.18	1.15
Parting		4.50	4.40	5.10
Karo III	10	5.50	3.25	7.66
Parting		6.00	4.40	8.60
Karo II	4	<1.00	0.17	1.25
Parting		4.00	4.00	5.00
Karo I	2		0.30	0.40

Parameters	Details
2.2.2.4 Description of Coal Seam	IS:
2.2.2.4.1 Seam Karo –VI to X	
All the Karo groups of seams occur in the order. Karo VI to X groups occurring in the exploitation in the block through opencas	e block and are designated as Seam Karo-I to X in ascending ne Barakar Formation is combined and thick and under active st.
2.2.2.4.2 Seam Karo –V	
Seam Karo-V is the bottom most Barakar deteriorated in quality as major part of th	r Formation. Seam Karo- V, though having workable thickness has e seam is composed of carbonaceous shale and shaly coal.
The borehole intersection of seam Karo- & 10 and CMESD-4,7,20,22 falling in carbonaceous shale.	V in boreholes MBD-5,6,7,14,15,16,17,18,20,24 & 25 and EBC-6 sector-II of Tisri Block shows that the seam is converted into
Thickness	
Thickness of Karo Seam-V varies from 0 has about 3.0 to 4.0m of thickness over Selected Dhori as most of the part has be been derived based on GR of Tarmi and before start of mining activity.	0.15m to 5.04m in Tarmi side and 0.44m to 4.90m in Tisri side. It er major part of the area. There is no information of Seam-V ir been converted to carb shale in this region. However, quality has d Tisri. Detailed analysis of Seam-V will be required in this zone
Roof and Floor	
The immediate roof of the seam is ge changes to grey shale which ultimately to Shale.	nerally medium to medium grained sandstone which gradually carbonaceous shell. The floor is mostly shale and carbonaceous

Dirt Bands

Number of dirt bands varies from 1 to 12 with a cumulative thickness varying from 0.15m to 4.85m. The seam is highly inter-banded in nature and there is wide variation in the percentage of dirt bands ranging from 45% to 98% of seam thickness. The bands are mainly represented by carbonaceous shell and shaly coal.

Quality

The grade of seam Karo-V in sector V of Tarmi part is G to Ungraded, whereas in Sector-II of Tisri part it is Ungraded. In Selected Dhori part the seam has mostly been converted to carbonaceous shale. Seam structure available for boreholes CDD-1, 2, 3, 4 and 9 only. In all boreholes except CCD-2 the seam is converted to carbonaceous Shale.

2.2.2.4.3 Seam Karo-IV

Detail with regard to seam thickness, quality and assessment of reserve of this seam is not available in the GR for Selected Dhori part.

In Tarmi block though its thickness varies from 0.10 to 1.30m, it is unworkable throughout the block excepting around bore holes EBC-7 & 22 (located wide apart) where its thickness is more than 1.0m. Quality characteristic for this seam is not available for Tarmi block. Though this seam has been established to be

	Parameters	Details					
	workable in a few bore holes but as such been assessed.	n no workable patch could be delineated, hence its reserve has not					
	In Tisri the thickness of the seam is usua II to W-II.	ally around 0.90m. The grade in Sector-II varies from Steel Grade-					
	2.2.2.4.4 Seam Karo-III						
	It is the Bottom-most workable seam among all the coal seams of the Karharbari Formation in the block. It has been encountered in 10 BHs drilled by GSI and CCL. It is completely burnt in boreholes EBC-11 & 21. It incrops in the northern part of the block and has roughly east-west strike.						
	Thickness						
	Thickness of Seam Karo -III varies from the area.	3.25m to 7.66m. It has about 5.0m of thickness over major part of					
	Roof and Floor						
	The immediate roof of the seam is gen sandy shale bands. The floor is mostly c	erally medium to coarse grained sandstone with intermittent thin coarse to gritty sandstone in nature.					
	Dirt Bands						
	The seam is almost free from dirt band. However, very thin band of carbonaceous shale has been observed in CDD-9 drilled by CCL.						
	Quality						
	In-band ash is available only in three boreholes CDD-1, 2 & 9 drilled by CCL. Accordingly, the ash varies from 23.8% to 28.0 %. In other boreholes drilled by GSI, EBC-7, 8, 9, 17 & 30; analysis shows ash percentage from 15.1 to 22.5 %. In major part of the area the grade of the coal is Steel-II to W-II. However, the grade becomes W-IV near CDD-1.						
2.2.3	Geological Block Area " Ha"	300					
2.2.4	Status of Exploration of the block	Explored					
2.2.5	Area covered by 'detailed' exploration within the block (sq. km)	-					
2.2.6	Whether entire lease area has been covered by 'detailed' exploration.	Yes					
2.2.7	No. of boreholes drilled within the block	Total (18+8) boreholes					
2.2.8	Whether any further exploration/study is required or suggested and time frame in which it is to be completed	A provision of 4-5 boreholes should be kept for drilling of about 500m to meet the emergent requirement of additional information at the time of mining coal from the property in the incrop zone.					
2.2.10	Overall borehole density within the block (no./ sq. km) approx	9					
2.2.11	No of Seams available as per GR (Geological Report)	Top to bottom					
		1. Karo VI to X combined					
		2. Karo V					
		3. Karo IV					
		4. Karo III					

	Parameters			Details						
					6	6. Karo I				
2.2.12	Seams not c Reasons	onside	red for Min	ing with	Sean due t	Seam IV, Seam II and Seam I. These seams are not mineable due to very low thickness.				
					Karo	VI to X combine	d has been cons	idered for minin	g. This seam	
					forms	s the topmost se	eam in the propo	osed mining are	a. The seam	
						peen mined exte	nsively with an c	occurrence of fire	e in the area.	
					Prese	ence of this sea	am is possible	in patches in t	he proposed	
					minin	ig area. Since its	s extent and quar	ntity could not be	e ascertained	
					due t	o past mining, it	ts reserve has n	ot been assesse	ed. However,	
					the s	ame needs to m	ined out along w	ith the underlyir	ng seams.	
2.2.13	Dip of the Se	am			The o	dip of ranges from	m 7-12°			
2.2.14				Seam	n wise	e Details of I	Reserves			
		Seam wise deta				eological Vis-à-v	vis Mineable Res	erves		
		SN	Seam Name	Thickr conside (m)) ered)	Thickness range (m)	Geological Reserve (MT)	Mineable Reserve (MT)		
		1	Karo VI to X	-		57.33-69.28	-	-		
		2	Karo V	2.0	0	2.92-4.93	7.01	4.51		
		3	Karo IV	-		0.18-1.15	-	-		
		4	Karo III	4.0	0	3.25-7.66	12.75	6.21		
		5	Karo II	-		0.17-1.25	-			
		6	Karo I	-		0.30-0.40	-			
			TOTAL		1		19.76	10.72		
2.2.15	Methodology of reserves estimation (also mention if any software package			MINEX software has been used for reserve estimation.						
2.2.16	Average Gra	dé					W III and G	<u> </u>		
2.2.17	Gross Geolo "Mte"	gical F	Reserve of t	he block	21.96					
2.2.18	Net Geologi "Mte"	cal Re	eserve of th	ne block			19.76			
2.2.19	Minable Res	erve o	f the block '	'Mte"			10.72			
2.2.20	Blocked Res	erve "I	Vite	some of			-			
2.2.21	the block "M	lte"	i actable Te				-			

	Parameters	Details
2.2.22	Percentage of Extraction	100%
2.2.23	Reserve already depleted (Base date of Mining Plan)	-
2.2.24	Balance Reserve (as on Base Date)	10.72 Mte

CHAPTER 3: MINING

	Parameters	Details							
3.1	MINING METHOD								
3.1.1	Existing method of mining if the mine is under operation	Opencast Mining with shovel-dumper combin	ation						
		Seams to be worked: Karo VI to X combined Choice of mining method: Considering the geo-mining conditions of the extraction of lower seams) viz.	l, Karo V and Karo III. Selected Dhori OCP (for						
3.1.2	Proposed method of mining with justification on suitability	 Gradient of seam floor, viz., 7° - 12° Multiple seams with variable thickness and 	I						
	of method of mining	Smooth flexible and apply operation	I						
		Smooth, flexible and easy operation The method of mining proposed to be adopted to extract coal and OB Selected Dhori OCM will be open cast mining with shovel-dumper combination of mining systems.							
		Life of mine: The mine life for nominal production is 8 years. The break-up of life of mine are as under: Particulars Years							
		Particulars	Years						
		Mine Life	8						
		Construction period	-						
		Opencast Mining with snover-oumper conditions of the seams to be worked: Karo VI to X combining Seams to be worked: Karo VI to X combining Choice of mining method: Considering the geo-mining conditions of the extraction of lower seams) viz. > Gradient of seam floor, viz., 7° - 12° > Multiple seams with variable thickness at > Smooth, flexible and easy operation The method of mining proposed to be adopt Selected Dhori OCM will be open cast mining Construction of mining systems. Life of mine: The mine life for nominal production is 8 year are as under: Particulars Mine Life Construction period Production build-up period Total period Total period Mine Boundaries: The mine boundary of the OCP has been fix East: A barrier of 30m has been kept betw and eastern limit of proposed quarry. West: The quarry surface of the western bor eastern bank of Tisri nala keeping a safe dia	3						
		Production period	3						
		Tapering period	2						
		Total period	8						
		 Mine Boundaries: The mine boundary of the OCP has been fixed as follows: East: A barrier of 30m has been kept between western limit of Tarmi (and eastern limit of proposed quarry. West: The quarry surface of the western boundary has been fixed along eastern bank of Tisri nala keeping a safe distance of 60m. 							

Parameters	Details							
	North:	The northern floor boundary ha	is been fixe	ed along the f	ault F7.			
	South:	A 45 m barrier between Gomo	oh – Barka	ikana loop lir	e and surface			
	edge of	the proposed quarry has been	kept.					
	Fina	al Stage Mine Parameters:						
	Param	eters	Unit	Minimum	Maximum			
	Dimen strike (sions of the quarry along on floor)	m	600	1300			
	Depth	of quarry	m	10	120			
	Dip ris	e length (on floor)	m	800	1450			
	Final C	uarry Floor area	Sq.km	1	.09			
	Final C	luarry Surface area	Sq.km	1	.79			
	Total ()R		66	2.12 2.00			
	Averee	uo Stripping Potio		6	16			
	Averaç Mine C	e Supping Raio		0	. 10 2 A			
	 Choice of technology: Considering the geo-mining conditions viz. Gradient of seam floor, v 12°, multiple seams with variable thickness and smooth, flexible an operation Shovel-dumper combination has been proposed for min quarry. Mining System & System Parameters As the seams are dipping moderate to steep, the mine will follow ho 							
	the mine The follo SI. No.	e. owing mining parameters have Particular	been cons	idered in the	project. /alue			
	1	OB Bench Height for 10-12 c shovels	um r	1	12-15			
	2	OB Bench Height for 5.5-6.5 cum shovels	m	1	8-10			
	3	Coal Bench Height for 5.5-6.4 cum shovels	5 m	1	8-10			
	4	Working bench width	r	1	40			

Parameters		Details 5 Non - Working bench width m 25 6 Bench Slope for OB and coal Deg 70 7 Blast Hole dia for OB and coal mm 250/160 8 Powder Factor in OB and coal Kg/cum 0.3-0.4 & 0.2 ine Scheduling and Calendar Programme of Excavation ine operation in the quarry can start from 1st year in (as the required land is ready in possession). Quarry will start from the northern side, south of fault 7 to touch seam-III floor at RL of 195m. Initially OB will be dumped in the dip de area. Working faces will advance towards south. As sufficient space is eated, internal dump leaving a distance of 60-100m from the working faces II start from the de-coaled area north to south. The OB dumped in the dip de in the initial period will be rehandled and dumped inside the quarry. he haul road layout of the mine has been shown in the final stage plan epared for the purpose. The width of the working and non-working benches as been kept at 40m and 25m throughout the life of the mine except first few ears when the size of the equipment permits for smaller working and non-prking benche width. Transporting of coal from the bench to the pit head coal ump will be carried out by hauling dumpers. The quarry will produce two prime will be carried out by hauling dumpers. The quarry will produce two prime will be carried out by hauling dumpers. The quarry will produce two prime transfer annual							
	5	Non - Wor	king benc	n width	m		25		
	6	Bench Slo	pe for OB	and coal	Deg		70		
	7	Blast Hole	dia for OE	3 and coal	mm		250/160	1	
	8	Powder Fa	actor in OE	3 and coal	Kg/cum	0.3	3-0.4 & 0.2		
	Mine So	cheduling a	nd Caleno	dar Programı	me of Exc	avatio	on	J	
1	Mine op	eration in th	e quarry c	an start from	1 st year in	(as the	e required land	l is	
á	already	in possessi	on). Quarr	y will start fror	m the north	nern si	de, south of fa	ult	
F	F7 to to	uch seam-III	floor at R	of 195m. Init	tially OB w	ill be d	umped in the c	qit	
5	side are	a. Working	faces will	advance towa	ards south	. As si	ufficient space	: iS	
	will start from the do coaled area porth to south The OB dumped in the din								
	side in the initial period will be rehandled and dumped inside the quarry								
	The hau	ul road layo	ut of the	mine has bee	en shown	in the	final stage pla	an	
۲ ۲	orepare	d for the pur	pose. The	width of the w	vorking and	d non-	working bench	es	
	has bee	n kept at 40	m and 25n	n throughout t	the life of the	ne mino	e except first fe	3W	
	vorkina	hench width	Transno	rting of coal fr	rom the her	nch to	the nit head co	n- nal	
	dump w	vill be carrie	d out by h	nauling dump	ers. The a	uarry	will produce to	wo	
	distinct	quality of c	oal throug	hout its life.	Based on	the n	ormative annu	ual	
	capacity	of the mine	e as 2.0 M	T, the propos	ed mining	sched	lule is generat	ed	
f	for 8 yea	ars of mine l	ife.						
-	The tar	geted coal j	production	from the mir	ne is envis	saged	in 4 th year. T	he	
	average	stripping ra	atio is 6.16	cum/te. The	summariz	ed mir	ning schedule f	for	
	coal extraction and corresponding overburden load for the project, annual								
(coal & OB production schedule has been provided in the tables below:								
	Summa	arized Minii	na Schedi	ule					
	Р	eriod	Year	Coal Production (Mt)	OE Remo	3 oval um)	Stripping Ratio		

Parameters			Details					
		1	0.50	3.	70		7.4	1
	Capacity built-up	2	1.00	7.	41		7.41	
		3	1.50	10	.18		6.7	9
		4	2.00	12.36			6.18	
	Production	5	2.00	11	.40		5.7	0
		7	2.00	6	.43 66		5.7	2 5
	Tapering	8	0.52	2	85		5.4	9
	TOTAL		10.72	66	.00		6.1	6
	The envisa	aged requirer dule	nents of HEMM	are	giver	n belov	W:	[]
	HEMM		Capacity	Ua	paci	y bui	u-up	Max
				1	2	3	4	
	OB							
	Diesel Hyd Shove		10-12 Cum	1	1	2	2	2
	Diesel Hyd Shove		5.5-6.5 Cum	1	2	2	2	2
	Rear Dumper		100 T	6	8	10	12	12
	Rear Dumper		60 T	6	8	9	10	10
	Diesel RBH Drill		250 mm	0	1	1	1	1
	Diesel RBH Drill		160 mm	1	2	2	2	2
	Dozer		410 HP	2	3	3	3	3
	Coal							
	Diesel Hyd Backh	oe	4.3-5.0 Cum	1	1	1	1	1
	Rear Dumper		60 T	2	2	4	4	4
	Diesel RBH Drill		160 mm	1	1	4	4	4
	Dozer with Ripper	Attachment	410 HP	1	4	4	4	4
	Common							
	Diesel Hyd Backh	oe	2-3 Cum	1	1	1	1	1
	FE Loader		5-6 Cum	1	1	1	1	1
	Dump Truck		20 T	2	3	4	4	4
	Grader		250-280 HP	1	1	2	2	2
	Wheel Dozer		460 HP	1	1	2	2	2
	R T Crane		40T	1	1	1	1	1
	R T Crane		20T	1	1	1	1	1
	Mobile Crane		8-14T	1	1	1	1	1
	Dozer with Ripper	Attachment	850 HP	1	1	1	1	1
	Tyre Handler		35 kN	1	1	1	1	1

Parameters		Details					
	Vibratory Compactor		1	1	1	1	
	Water Sprinkler	28KL	1	2	2	3	3
	Reclamation						
	Water Sprinkler	28 kL		1	1	2	2
	Dozer	410 HP		1	1	2	2
	Mining Method for Overburden The OB benches along with major different shovel dumper configur proposed to be excavated by 10 RD. Thinner partings are propose hydraulic shovels with 60T RD. F	Removal parting are pro- ations. The to -12 cum diese ed to be exca or the estimation	opos p OE I hyd avate on o	ed to 3 and drauli ed by f the	be ta thick c sho 5.5-(dump	ken us < partir vel wit 6.5cum er pop	ing two igs are h 100T diesel ulation,
	the lead for OB transportation has bench.	been consider	ed fo	or ead	ch yea	ar and fo	or each
	Mining Method for Coal Winning						
	For coal production, 4.3-5.0 cum have been proposed.	diesel hydrauli	c ba	ckho	e with	60T R	D
	Drilling & Blasting						
	The drilling and blasting opera necessary before excavation by provided in the project will carry of will be done in patterns decided in and as per the conditions laid dow	tions for loos shovels. The ut the site prep advance depe vn by DGMS.	ening suffi arati endin	g of icient on. T ig on	coal num he bla the st	and (ber of asthole rata ha)B are dozers drilling rdness
	Blasthole drills of 250/160 mm dia and coal benches.	meter will be u	sed f	or dr	illing iı	n OB/ p	artings
	The standard practice involving detonating cord, detonating rela Heavy ANFO, slurry or emulsion e for blasting.	the electric de ys to achieve xplosives as th	etona hole ie co	ators e-to-ł lumn	for th nole d charg	ne initia Ielays, ge will b	ition of use of e used
	Seam Wise Extrac	table Reserv	es (i	n MT)		
	Name of Seam	Extrac	table	Res	erve ((MT)	
	VI to X Combined			-		. <i>.</i>	
	V		4	1.51			
			(5.21			
	Total		1	0.72			

	Parame	eters				Details			
			Dumpi	na Stra	ateqv				
			Total or of OB. rehand the qua surface on eac maximu	verburd Out o led fror arry itse e level. h deck um RL	len quantities es f which about § n 4 th year onwar If. The proposed Total two decks t have been pro of 270m.	timated for Sel 9.72 Mcum of ds. All the OB dump height is of height 30m posed and the	ected Dhori O(OB has beer removed will b s maximum 60 n each leaving e final stage c	CM is 66. propos e backfill m from ir 30m wid ump will	00 Mcum ed to be led within nmediate de berms I attain a
			The internal dumping will be carried out from the benches in the batte Dumpers movement will be so planned as to avoid the unnecessar movement of dumpers in different horizons.					ne batter. necessary	
			To maintain the stability of the internal OB dump, the precautions laid down in Chapter 4 should be strictly adhered. Regular monitoring of bench, quarry high walls and spoil dumps stability should be done using Slope Stability Radar or similar instrument.					laid down h, quarry Stability	
		The Tentative details of internal and external dumps OB quantity is give below:				is given			
					Dump	Volume (Mo	cum) Top	RL (m)	
					External	-		-	_
					Internal	66.00	+	270	
					Total	66.00	+	270	_
3.1.3	Coal production	on capacity				2.0 Mtpa			
3.1.4	Justification for Coal production	optimization capacity	Coal p normat	roducti ive anr	on capacity ha nual advance.	is been fixed	based on co	al thickr	ness and
3.1.5	Calendar year fr	om which the	1 st Year						
3.1.6	Year of Achi	eving rated				4 th Year			
3.1.7	production Tentative Coal Production Plan "Mte" .7								
		Period		Year	Coal Production (Mt)	OB Removal (Mcum)	Stripping Ratio (cum/te)		
l			-up	I	0.50	3.10	1.41		

	Parameters			Details		
		2	1.00	7.41	7.41	
		3	1.50	10.18	6.79	
		4	2.00	12.36	6.18	
	Production	5	2.00	11.40	5.70	
		6	2.00	11.43	5.72	
	Tapering	7	1.20	6.66	5.55	
	Tupering	8	0.52	2.85	5.49	
	ТО	TAL	10.72	66.00	6.16	
3.1.8	Rated Capacity "Mtpa"					
	- By OC			2 Mtpa		
	- By UG			-		
	- Overall			2 Mtpa		
3.1.9	Life of the mine: "Years"					
	- By OC			8 Years		
	- By UG			-		
	- Overall			8 Years		
3.1.10	Whether the proposed					
	external OB dump site is					
	coal/ lignite bearing: If so,			Not applicable	9	
	whether coal/lignite below					
	waste disposal area is					
3111	Whether pegative proving for					
5.1.11	coal / lignite in the proposed					
	site for OB dump/			Not Required		
	infrastructure has been done.					
3.1.12	Results of any investigation					
	carried out for scientific	Slope stability	study for pit an	d dump slopes.	hydro-geology stu	idy and
	mining, conservation of	washability stu	ldv proposed	, <u>1</u> ,	, <u> </u>	5
	minerals and protection of					
	environment; future					
	proposals.					

CHAPTER 4: SAFETY MANAGEMENT

	Parameters	Details
4.1	Safety Management	
4.1.1	.1 Important safety aspects: Major Risks and	Safety of men and machine deployed in the mining area should be properly taken care of irrespective of whether the mining activities are performed by departmental or by outsourcing means.
	viz. Proximity to river, adjacent working, geo- mining disturbances, slope stability and remedial measures suggested.	All the the statutory provisions laid down in The Mines Act 1952, Coal Mine Regulation 2017 and specific permission from DGMS relating to mining in general and opencast mining in particular have to be adhered to and implemented in order to maintain day to day safety.
	It should also include	1. Safety aspects for of HEMM / equipment
	proposed overall slope of the quarry and OB dump, dump height, strata control, fire and spontaneous heating, gas monitoring, disaster management,	Special precaution should be taken while deploying workers in the mine. Before employing any person to the mine proper vocation training should be imparted and recommendations of various Safety Conferences should be strictly followed. Some of the major aspects are as follows: -
	danger from inrush of water	A) For persons:
	eic.	i) No persons shall be deployed unless he is trained at VTC and holds VTC Certificates. A record of the same shall be maintained.
		ii) Records in Form-B and Form-D shall be maintained.
		iii) Records of driving license of operators shall be kept by competent authority and shall be made readily available for inspection by management.
		iv) Adequate supervision shall be maintained by competent persons, including officials and technicians.
		B) For Machineries: Provisions of Regulation 109, 110, 216 & 217 of CMR 2017 and DGMS Cir. (Tech.) 1 of 1999 should be strictly adhered to along with the following:
		i) All machinery and plant used in connection with working of a mine shall be of good design, sound construction, and suitable material, adequate strength, free from patent defect and properly maintained.
		ii) The owner, agent and manager shall provide adequate training facilities and ensure proper training of persons employed for operation and maintenance of machinery and plant.
		iii) No person except an engineer or other competent person under his supervision shall undertake any work on machinery and plant in which technical knowledge or experience is required.

Parameters	Details
	iv) All the machineries to be deployed in mines shall be so designed as to afford the operator clear and uninterrupted vision all around.
	v) Every heavy earth moving machineries, including trucks and tippers, used in mine shall be fitted with adequate safety features or devices as specified by DGMS. All equipment shall be provided with audio-visual alarms, proper light for use at night and fitted with suitable type of the fire extinguishers.
	vi) Truck mounted drill machines designed for tube well drilling for sources of water shall not be used and only proper type of blast hole drill machine, especially designed for mining purpose, shall be used in the mine.
	vii) Every heavy earth moving machinery shall be under the charge of a competent person (Operator or Driver), authorized in writing by the Manager.
	viii) All persons employed or to be employed to operate heavy earth moving machinery shall be trained and their competency shall be evaluated by a Board constituted by the management, who shall be persons who are not connected with imparting of training.
	ix) A proper record of repair and maintenance along with inspection done by competent authority and defect pointed out shall be maintained and signed by authorized person.
	x) Only such fitters or mechanics possessing driver's or operator's license, shall be allowed to carry out test-run of heavy earth moving machineries.
	xi) No person other than the operator or the driver or any person so authorised in writing by the manager shall be allowed to ride on a heavy earth moving machinery
	C) General:
	i) Every person shall strictly adhere to the provisions of the Act and of the rules and regulations and to any order or direction issued by the manager or an official with a view to the safety or convenience of persons not being inconsistent with the Act, rules and these regulations; nor shall he neglect or refuse to obey such orders or directions.
	ii) Every person shall, immediately before proceeding to work and immediately after terminating work at the end of his shift have his name recorded in the appropriate register.
	iii) Risk Management Plan of tipper/pay loader shall be made and implemented.

Pa	arameters	Details
		iv) All operators/drivers so authorised by the Manager shall observe the Regulation 62 and 63 of CMR 2017 and obey the systematic traffics rules prepared by management
		v) Before deploying workers, they must be trained and briefed about safety aspects in opencast mine. However, during course of execution of the work, if any accident occurs whether major or minor, the matter shall have to be immediately informed to mine management i.e. Colliery Manager/Agent/GM of Area so that Notices of accidents in a accordance of (Reg. 8 of CMR 2017) and Section 23 of The Mines Act 1952 may be given and other necessary steps may be taken in accordance with the Mines Act 1952.
		vi) Mine authority shall operate transport system in such a way so as to minimize pollution in the mine.
		2 Stability of Benches, Quarry Highwalls and Spoil Dumps
		During quarry operations, it is necessary to adopt required mining parameters for the stability of benches, highwalls and spoil dumps. It is also mandatory to examine systematically the fencing of mine workings, landslides and cracks between benches. It is required to maintain well- graded and wide roads on benches keeping the width of working areas sufficient for spreading of blasted rock and movement of the mining and transport equipment.
		During actual mining operation, systematic observations of the condition of benches, high wall slopes and spoil dumps should be carried out and the dimensions be modified if necessary to suit the local conditions. To ascertain the optimum slope angles for stability of quarry benches, highwalls and spoil dumps, scientific study of slope stability along with hydro-geological study of the area needs to under taken.
		During actual mining operation, systematic observations of the condition of benches, high wall slopes and spoil dumps should be carried out and the dimensions be modified if necessary to suit the local conditions.
		Provisions laid down in Reg. 106 and 108 of the Coal Mines regulation 2017 shall be strictly adhered to for the safety of quarry and OB/ spoil dumps. In addition to this, the following precaution should be considered:
		i) The spoil dump height should not exceed 90m from immediate surface level with an overall slope of 28 ^o or less. In the event of encountering steep floor gradient, floor blasting should be done and the area properly levelled by dozer before spoil dumping.

Parameters	Details
	ii) No working or construction should be allowed within the 60m toe of the OB dump.
	iii) Before dumping the OB on the floor of seam, at least 10m length all along the strike length should be made horizontal at every 50 meter by floor dinting/blasting.
	iv) Dump should be created in such a way that there is no chance of accumulation of water in and around the base of dump as it will adversely affect the shear strength of the base material of dump. It must be ensured that there is no stagnant water at the toe of dump and the top of the dump.
	v) The toe and face of the dump should not be eroded or cut at any point of time to avoid slope failure. A suitable toe wall should be created along the dump periphery.
	vi) Formation of dumping should be done in square or circular or any regular shape as far as possible.
	vii) Proper drainage system should be provided to bring down rain water by construction of inclined drain on dump face and catch drain on all benches.
	viii) During active period of dump, all rain water should be diverted away from mining site as far as possible.
	ix) Sump and pumping capacity should be sufficient to accommodate peak surface run-off and seepage of water.
	x) Gabion wall and garland drain should be constructed and maintained to trap the surface run-off and sludge coming from dump.
	xi) Plantation and grassing should be done on top and slope of the dump respectively.
	xii) Regular monitoring is required for development of tension crack, gullies, movement of soil mass, stagnation of water and any other unusual occurrence. In case of dump movement, rate of movement of dump should be monitored. Special attention should be given at curve area/turning area of the dump.
	3 Precautions Against Danger of Inundation from Surface Water
	i) Adequate protection against any danger of inrush of surface water into the mine or part shall be provided and maintained to the satisfaction of DGMS, whose decision shall be final.
	ii) The entrance into the mine shall be so designed, constructed and maintained that its lowest point (which means the point at which a body

Parameters	Details
	of rising water on surface can enter the mine) shall be not less than 3.0 meters above the highest flood level at that point.
	iii) Every year, during the rains constant watch shall be kept on the flood levels on the surface of the mine and if at any time the levels cross the highest levels earlier recorded, such levels shall be marked by permanent posts along the edges of water and the new highest levels thus observed shall be recorded with the date as the highest flood level on the plans by an actual survey.
	iv) If water dams or reservoirs are built across rivers and water courses on the upstream side of the mine, arrangements shall be made for communication between appropriate authorities for the purpose of ascertaining the quantity and timing of water released from the dams which is likely to endanger safety of the mine and arrangement for similar communication shall be made when water level rises on the upstream side which is likely to endanger the mine.
	v) The highest flood levels and danger levels at least 1.2 meters below the highest flood level, shall be permanently marked at appropriate places on the surface and whenever water rises towards the danger level at any place, all persons shall be withdrawn from the mine sufficiently in advance and for this purpose adequate arrangements of quick communication to all parts of the mine by effective systems shall be provided and maintained.
	vi) No working shall be made in the mine at any spot lying within a horizontal distance of 15 meters from either bank of a river or nala.
	vii) A competent person shall, once at least in every fourteen days during the rainy season and once at least in every thirty days during other periods of the year, examine every protective measure provided under regulations 149, whether in use or not, for their stability, and a report of every such examination shall be recorded. The protective measures and workings shall also be inspected, once at least in every quarter by the Manager personally.
	viii) A careful assessment is to be made against the danger from surface water before the onset of rainy season. The necessary precautions should be clearly laid down and implemented. A garland drain needs to be provided to drain away the surface rainwater from coming into the mine.
	ix) An embankment, 3.0m above the HFL, along the Bokaro River and Nakti nala should be made. Inspections for any accumulation of rainwater, obstruction in normal drainage and weakening in the embankment should be made.

Parameters	Details
	x) Standing order for withdrawal of working persons in case of apprehended danger. During heavy rain inspection of vulnerable points is essential. In case of any danger persons are to be withdrawn to safer places for Protection of Equipment Deployed at bottom horizons from flooding.
	During the heavy monsoon period, the mining operation in the lower- most bench may have to be stopped. Therefore, it is proposed to drown the lower-most bench, which would work as a sump. The water will be pumped out and discharged into the nearby nala/ river after proper sedimentation.
	For ensuring safety of the equipment while working out bottom horizons with no access to surface profile, the following measures should be taken:
	i) Drivage of initial trenches if any and coal cutting on bottom benches should be done during the dry period of the year.
	ii) Ramps should be made for quick shifting of equipment from bottom horizons, liable to be flooded during monsoon period, to the top horizons.
	5 Prevention of Electric Shocks:
	During mining operations, all the statutory provisions of the Indian Electricity Rules 1956, and Indian Standards for installation and maintenance of electrical equipment etc. should be observed.
	i) For protection from electric shocks to persons, all electrical equipment with voltage up to 1000V should be provided with Earth Leakage Relay, which will automatically disconnect electrical circuits.
	ii) Closed mobile substations and switchgears should be mechanically interlocked which exclude the possibility of opening the door when oil switch and air circuit breakers are in operation.
	iii) All metal parts of electrical equipment should be properly earthed to avoid failure of insulation.
	iv) All H.T lines and cables located within the blasting zones should be disconnected during charging & blasting operations.
	6 Dust Suppression & Dilution of Exhaust Fumes:
	For precaution against dust, Regulation 143, 144 and 145 of CMR 2017 should be observed. Beside this the following measures should be adopted for dust suppression at all quarry working places, dumps, haul roads, CHP and near other auxiliary mining operations.

Parameters	Details
	i) Spraying with water on all working faces & haul roads, by special spraying machines or water-sprinkler.
	ii) While drilling holes, it is necessary to use dust extraction devices.
	iii) Installation of local dust suppression and air conditioning devices in cabins of excavators and drilling rigs may be considered.
	iv) Leveling of spoil dump surface.
	v) Separate dust suppression arrangement should be provided for CHP.
	To prevent collection of harmful mixtures in the atmosphere, from the different sections of quarry workings, it is recommended: -
	To spread out the sources of dust formation and omission of harmful gases throughout the working area of the quarry, the following precautions should be taken:
	i) Drilling & blasting operations should be timed for periods of maximum wind activity during the day.
	ii) Dumpers may be provided with purifiers for exhaust gases.
	7 Measures to Be Taken for Fire Fighting and Fire Prevention:
	In addition to statutory provisions as laid down in Reg 135, 139 and 140 of CMR 2017, the measures for firefighting and prevention of fires are as follows:
	i) Organisation of special cell for systematic observations to examine and prevent fire.
	ii) Removal of spillage of coal on benches and cleaning of coal horizons to prevent cases of coal heating.
	iii) Storage of lubricants and cotton waste in enclosed fireproof containers in working places.
	iv) Provision of fire extinguishers.
	8 Measures to Be Taken While Working Above Underground Galleries:
	In addition to provisions laid down in DGMS Circulars (Tech. 2 & 3 of 1980, Tech. 11/1979), the additional measures for extracting pillars by opencast method are as follows:
	 i) Quarry shall be worked by Heavy Earth Moving Machinery only. No manual operation in the quarry will be done.

Parameters	Details
	ii) HEMMs, except drilling machines shall not be deployed on the bench where thickness of coal or overburden above the UG galleries, as proved by advance boreholes or other suitable methods, is less than 6m.
	iii) Exposed coal faces (including UG galleries shall be kept covered with fine grained incombustible OB material to prevent breathing of air and control fire to dip side working. This cover shall be removed only at the time of coal extraction.
	 iv) Overburden containing carbonaceous material shall not be dumped within 30m of the exposed side of the coal benches. Hot overburden shall be quenched and cooled at dump sites.
	v) No person shall be allowed at any place in the opencast working where the thickness of overburden and/or coal over any gallery is less than 1.5m.
	vi) Except for the purpose of inspection and support work no person shall be allowed in the underground mine beneath and within 200m of the opencast excavation. The person visiting UG will take all safety precautions for safe working.
	vii) Blasting in fire area
	 No explosive other than slurry and emulsion explosive shall be used.
	 Blasting shall be done with detonating fuse down the hole. Fresh drill holes should be tightly plugged at the mouth.
	• Temperature inside the hole shall be measured by bi-metallic thermocouple heat sensor (before filling with water) and if the temperature exceeds 80°C in any hole, the hole will not be charged.
	• All blast holes shall be kept filled with water. When any hole is traversed by cracks or fissures the hole shall not be charged unless it is lined with an asbestos pipe and the hole filled with water. In addition, bentonite should be used for sealing any cracks at the bottom of the hole.
	 Detonating fuse shall not be laid on hot ground without taking suitable precautions.
	 Charging and firing of holes in any one round shall be expeditiously completed and in any case within 2 hours.
	• A parting of at least 2m between the bottom of a short hole and roof of underground gallery shall be left intact.
	• Effective muffling of hot shot holes with old wire rope screens shall be done for prevention of flying hot fragments.

	Parameters	Details
		No blasting shall be done in crushed or broken ground.
		 No person shall be employed within 150m when blasting the heated material.
		• The spacing of hole in the coal/OB benches lying immediately above the galleries shall be so adjusted that the holes do not lie immediately above the galleries in order to ensure that blast holes do not directly fire into the underground working.
		• All holes in the coal/OB benches lying immediately above the galleries shall be charged with water impulses or with moist sand of at least 0.6m in length at the bottom of the hole.
		 No person including a shot firer shall take shelter within 100m of the quarry opening. Such shelter shall be of an approved design.
		9 Measures to Be Taken While Drilling Blasting:
		Following measures should be taken during drilling and blasting operation in the quarry beside the statutory requirements:
		i) Drilling and Blasting in quarry should be done in accordance with the provisions of Mines Act, rules and regulations and based on the Standing Orders for the safe use of explosives.
		ii) Adequate safety measures have to be taken during blasting operation in the quarry so that men/machine are not affected.
		10 Conservation
		Suitable measures should be taken to minimize coal loss during mining operations. Selective mining of in-seam dirt bands has been proposed. It is proposed not to dump any spoil material over coal bearing area, amenable for mining, at present or even at a future date.
		11 Scientific Studies
		The slopes of the quarry and dumps have been proposed on the basis of experience in the adjoining areas. However, to ascertain optimum slope angles for stability of quarry batter and dumps a scientific study need be carried out. Similarly, hydro-geological study of the area is to under taken as none is available at present. Studies should also be carried out to ascertain the pattern of surface drainage, the manner of diversion of water courses to other water courses away from the mining area and the dimension of diversion dams, garland drains and other protective structures to be constructed.
4.I.Z	Company Board that entire mining operation will be carried out as per the	

Parameters	Details
Statutory provision given under Mines Act 1952, Coal Mine Regulation 2017 and & wherever specific permission will be required the company will approach the concerned authorities.	

	Parameters	Details
5.1	Mine infrastructure required e.g. Equipment maintenance planning, Office buildings, Workshop.	The life of Selected Dhori OCM has been estimated as 8 years. For service and welfare buildings, temporary type of construction has been proposed. Provision for service buildings including PO office, pit office, magazine and other statutory buildings has already been proposed.
5.2	Power supply & illumination	It is envisaged that this project will receive power at 33 kV from Bokaro Thermal Power Station (BTPS) of DVC. For this, two nos. existing 33 kV overhead lines from BTPS to B&K Main-sub-station shall be extended up-to proposed quarry. A 33 kV energy meter shall be installed for recording its energy consumption. Approximate distance of the quarry from B&K Sub-station is approximately 5 km. Further, it is proposed to establish 1 no 2 X 5 MVA, 33/6.6 kV sub -station with provision for 2 nos. incoming 33 kV feeders and 12 nos. outgoing 6.6 kV feeders for supply of power to different power consuming equipment of the project.
		Haul Road Illumination
		For illumination of permanent haul roads, 250 W HPSV lamps and luminaires mounted on 11/13 m (approx.) high steel poles have been envisaged. Temporary haul roads shall be illuminated according to the position of working and available facility.
		Illumination of Quarry General Area/ Dump Area
		Illumination of quarry general area/dump area will be done with 400W HPSV lamp in flood light fixtures mounted on lighting towers. Sufficient no. of 25/10 kVA, $6.6 \text{ kV} / 0.23 \text{ kV} (L - L)$ transformers have been provided for illumination of quarry general area / dump area / haul road / face etc.
		Service Road Illumination
		150 W HPSV lamps in street light luminaires are proposed for service road illumination in the project. The luminaires on steel tubular poles 11/13 m high is proposed for this purpose.
5.3	Drainage & Pumping:	During the rainy season, water will be allowed to accumulate in the sump on the floor of lower most seam of the quarry. Sumps are provided for the dewatering pumps which will transfer in-pit water for controlled discharge off the site. The planning of dewatering the mine shall be done in such a way that the working faces and haul roads in the quarry shall remain dry as far as possible. The layout of the quarry provides suitable gradient along the quarry floors and the benches to facilitate self-drainage of water to the sump at the lowest level of the quarry. Water accumulated in the mine sump will be pumped out to the surface.

	Pumping Capacity	Probable water accumulation on the day of maximum rainfall taking 10% for seepage and underground precipitation = 144082 Cum. Pumping capacity/hr = 1201 Cum/hr (333 lps)
	Pump Selection	Main Pumps- Two no of 120 lps x 180 m head pumps have been provided for the mine.
		Low Head Pumps
		Besides the above main pumps one no of 75 lps x 80 m head pumps have also been provided to dewater the quarry during mining operation. Face Pumps and Slurry Pumps
		Two no of Face Pumps of 15 lps. x 60 m head and one no of Slurry Pump of 22 lps x 45 m head have been provided to pump out the water & slurry respectively accumulated near the working faces of the quarry. Diesel Pump
		One number of 75 lps x 180 m head diesel pump has been provided for emergency requirement.
5.4	Coal Handling Arrangement: Brief detail of the CHP/ Mode of Dispatch, Coal quality and Coal staking and handling arrangement	The total production of the mine has been the proposed as 2.0 MTY. The existing arrangement of coal handling will be utilised.

CHAPTER 6: LAND REQUIREMENT

	Parame	ters				D	etails			
6.1	LAND REQUIRE	EMENT	•							
			Break	up of p	ore-minii	ng land t	ype (indic	ative) a	nd sour	ce of data.
					Ту	/pe of Lar	nd		Area in I	Ha.
						Tenancy				
6.1.1	Total Land requi mine in "Ha"	rement for the)		Gov	⁄t Non-Fo	rest		Nil	
						Forest			211.8	2
						Total			211.8	2
			Source-	Plan sı	upplied by	y Project.				
	During mining La	and use detail	s (Area in F	la):						
	Туре	Land use (Proposed)	Land Use (End of Life)	Agri cul tura I	Plant ation	Water Body	Public/ Compa ny Use	Forest Land (Retur ned)	Undi st urbe d	Total
	Europeanties Area	179.44		land						
	Excavation Area		151 51		151 51					151 51
	Excavated Void		27.93		101.01	27 93				27.93
	Without plantation		21.00			21.00				21.00
	Top Soil Dump									
C 4 O	External Dump									
0.1.Z	Safety Zone									
	Haul Road									
	Road diversion									
	Diversion/holow									
	River/Nala/canal	1.48	1.48			1.48				1.48
	Settling pond									
	Road & Infrastructure area	5.06	5.06				5.06			5.06
	Rationalization									
	Garland drains									
	Embankment	1.69	1.69		1.69					1.69
	Green Belt	24.15	24.15		24.15					24.15

	Parame	eters				D	etails				
	Water Reservoir near pit										
	UG entry Undisturbed/ Mining right for UG										-
	Resettment Pit head power plant										-
	Water harvesting										
	Agricultural land										_
	Total	211.82	211.82		177.35	29.41	5.06			211.82]
			1								
6.1.3	Surface feature block area	es over the	North-so flowing D	uth flo [.] Damod	wing Tisr ar River.	i Nala & 、	loria Nala	dischar	ging to '	West to e	ast
6.1.4	No. of villages/ shifted	Houses to be	Nil								
6.1.5	Proposed programme	Rehabilitation	No R&R	involv	ed in this	Project					

 Commitment from the project proponent that the company will comply Environment and Forest Condition stipulated in the respective clearances 	In order to carry out the proposed mining activity in an environmentally sustainable manner, suitable environmental protection measures shall be taken up at different stages of project operation and post closure In addition to this, few environmental protection measures have been suggested as a part of mine closure activities.
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CHAPTER 8: PROGRESSIVE & FINAL MINE CLOSURE PLAN

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	SI. No	Land use During Mining	g		Post Mining Land	Use Pla	n
		Particulars	Area (Ha.)		Particulars		Area (Ha.)
	1	Quarry	179.44	Interna Planta	al dump reclaimed tion	d with	151.5′
					void filled with wate	r	27.93
	2	Infrastructure (CHP, Colony, Road)	5.06	Infra	astructure for Future	Use	5.06
	3	Green Belt/Safety Zone	24.15	Planta	ation on Green Belt/ Zone	Safety	24.15
	4	Tisri Nala and Joria Nala	1.48	Ti	sri Nala and Joria Na	ala	1.48
	5	Embankment against nala	1.69	Pla	ntation on Embankn	nent	1.69
		Total Project Area	211.82		Total Project Area	l	211.82
1 г					closure of the fill	ne	
	S. No.	Activity			Weighted % of M	ne ine Clos	sure Co
	S. No.	Activity			Weighted % of M Progressive	ne ine Clos F	sure Co
	S. No.	Activity Dismantling of Structure			Weighted % of Mi Progressive	ne ine Clos F	sure Co
	S. No.	Activity Dismantling of Structure Service building			Weighted % of M Progressive	ne ine Clos F 8	sure Co inal 3.50
=	S. No.	Activity Dismantling of Structure Service building Residential Building			Weighted % of Mi Progressive 0	ne ine Clos F 8	sure Co Final 3.50
	S. No.	Activity Dismantling of Structure Service building Residential Building Industrial Structure			Weighted % of Mi Progressive 0	ne Clos F	sure Cc Final 8.50
	S. No. A B	Activity Dismantling of Structure Service building Residential Building Industrial Structure Safety & Security			Weighted % of Mi Progressive 0	ne Clos F	sure Co Final 3.50
	S. No. A B	Activity Dismantling of Structure Service building Residential Building Industrial Structure Safety & Security Random rubble mansory/conceree Tase well around dump (Oabbien	ete wall		Weighted % of M Progressive 0	ne Clos F	sure Co Final 8.50
	S. No. A B	Activity Dismantling of Structure Service building Residential Building Industrial Structure Safety & Security Random rubble mansory/concere Toe wall around dump/Gabbion v Barbared wire fencing	ete wall wall		Weighted % of M Progressive 0 6.50	ne ine Clos F 8	sure Co Final 3.50 3.20
	S. No.	Activity Dismantling of Structure Service building Residential Building Industrial Structure Safety & Security Random rubble mansory/concerent Toe wall around dump/Gabbion v Barbared wire fencing Fencing/boundary wall, fencing a	ete wall wall	er body	Weighted % of Mi Progressive 0 6.50	ne ine Clos F 8	sure Co Final 3.50 3.20
	S. No. A B	Activity Dismantling of Structure Service building Residential Building Industrial Structure Safety & Security Random rubble mansory/concere Toe wall around dump/Gabbion v Barbared wire fencing Fencing/boundary wall, fencing a Garland drains	ete wall wall around wate	er body	Weighted % of M Progressive 0 6.50	ne ine Clos F 8	sure Co Final 3.50 3.20
	S. No. A B B	Activity Dismantling of Structure Service building Residential Building Industrial Structure Safety & Security Random rubble mansory/concere Toe wall around dump/Gabbion v Barbared wire fencing Fencing/boundary wall, fencing a Garland drains OB Dump Reclamatiom	ete wall wall around wate	er body	Weighted % of M Progressive 0 6.50	ne ine Clos F 8	sure Cc Final 3.50 3.20
	S. No. A B B C C A	Activity Dismantling of Structure Service building Residential Building Industrial Structure Safety & Security Random rubble mansory/concere Toe wall around dump/Gabbion v Barbared wire fencing Fencing/boundary wall, fencing a Garland drains OB Dump Reclamatiom Technical Reclammation	ete wall wall around wate	er body	Weighted % of Mi Progressive 0 6.50	ne ine Clos F 8	sure Co Final 3.50 3.20
	S. No. A B B C C A	Activity Dismantling of Structure Service building Residential Building Industrial Structure Safety & Security Random rubble mansory/concere Toe wall around dump/Gabbion v Barbared wire fencing Fencing/boundary wall, fencing a Garland drains OB Dump Reclamatiom Technical Reclammation Re-handling of OB	ete wall wall around wate	er body	Weighted % of M Progressive 0 6.50 60.50	ne ine Clos F 8	sure Co Final 3.50 3.20

1.0	Dump		
	Biological Reclammation & Plantation		
	Top soil Management		
	Grassing of OB dump		
	Planatation around virgin Area, safety zone, green	15.00	11 7
	external Dump and internal reclaimed area	10.00	
	Plantation post care (including manpower)		
	Plantation over cleared area obtained after dismantling		
D	Land scaping of the open space in leasehold area		
-	for improving its		
	esthetic. Drain, Pipe lines, Peripheral road, gates,	4.00	5.50
	Viewpoints, cemented steps on bank		
	Development of Agriculture land		
E	Air Quality (Water tanker Sprinkler & ether		
	Control measures)	12.00	1.50
	Water Quality (FTP & STP etc operating cost)		
	Manpower Cost and supervision		
F	Post Closure Menitoring		
F	Post Closure Monitoring		
F	Post Closure Monitoring Air Quality	0.00	3.20
F	Post Closure Monitoring Air Quality Water Quality Power Cost	0.00	3.20
F	Post Closure Monitoring Air Quality Water Quality Power Cost Manpower Cost and supervision	0.00	3.20
F	Post Closure Monitoring Air Quality Water Quality Power Cost Manpower Cost and supervision	0.00	3.20
F	Post Closure Monitoring Air Quality Water Quality Power Cost Manpower Cost and supervision Entrepreneurship Development (Vocational/skill development	0.00	3.20
F	Post Closure Monitoring Air Quality Water Quality Power Cost Manpower Cost and supervision Entrepreneurship Development (Vocational/skill development training for sustainable income of affected people)	0.00	3.20 0.50
F G	Post Closure Monitoring Air Quality Water Quality Power Cost Manpower Cost and supervision Entrepreneurship Development (Vocational/skill development training for sustainable income of affected people) Miscellaneous & Other measures like Colden	0.00	3.20
F G	Post Closure Monitoring Air Quality Water Quality Power Cost Manpower Cost and supervision Entrepreneurship Development (Vocational/skill development training for sustainable income of affected people) Miscellaneous & Other measures like Golden Handshake one time	0.00	3.20 0.50 5.40
F G H	Post Closure Monitoring Air Quality Water Quality Power Cost Manpower Cost and supervision Entrepreneurship Development (Vocational/skill development training for sustainable income of affected people) Miscellaneous & Other measures like Golden Handshake, one time financial grant, alternative jobs, other services etc.	0.00	3.20 0.50 5.40

		Selected
	Project Name	Dhori(lower)
	Project Area (Ha)	211.82
	Escrow Amount per Ha. For OC Project as on April, 2019 (lakhs/ Ha)	9
	WPI as on April 2019	121.10
	WPI as on April 2021	132.00
	Escrow Amount per Ha. For OC Project as on April 2021(lakhs/ Ha)	9.81
	Current value of corpus as on April 2021 (lakhs)	2077.97
	Amount deposited till 31.03.2021 (lakhs)(in mine closure account of SDQ-1 & SDQ-3	2453.52









