



Classification of Ni Levels for Determination Cut-Off Grade in Region X

Nurliah Jafar^{1*}, Hendra Wahid R², Sri Widodo³

^{1,2} Department of Mining Engineering, Faculty of Industrial Technology, Universitas Muslim Indonesia, Indonesia

³ Department of Mining Engineering, Faculty of Engineering, Universitas Hasanunndin, Indonesia

*Correspondence e-mail: nurliah.jafar@umi.ac.id

ABSTRACTS

The need for ore with levels above 1.8% presents the author's initiative to conduct research to produce ore products with the required levels. The aim is to find out the lowest grade of ore that can be mined and used as a cut-off grade in area x. The research method used is the blending method. Some of the data needed include the drill point coordinates, the borehole's depth, the borehole, the slope of the borehole, lithology, and the value of Ni content. The distribution of nickel in the x region is dominated by ore which has a grade of 1 to 2% with a total volume of 140,100 Bcm, and there is a small amount of nickel that has a rate above 2% with an importance of 15,650 Bcm. The mining sequence has been divided into three to three stages, where the tonnage obtained from each location is 8,9937 tons, 442361 tons, and 100,554 tons. The simulation of blending steps was carried out starting from the classification of nickel with the highest grade, namely nickel with the lowest rate of 1.8% with an average rise of 2.07%, to the last stage, which produces a middle grade of 1.9%, namely in the classification of nickel with rates above 1.6 %. From the results of the blending simulation that has been carried out, it can be concluded that the lowest grade that can be used as a cut-off grade in area x is ore with a Ni content of 1.6%.

ARTICLE INFO

Article History:

Received 03 April 2022

Revised 05 April 2022

Accepted 27 June 2022

Available 30 June 2022

Keyword:

Ores; Cut-Off Grade;

Volume; Tones; Blends

© 2022 Journal of Geology & Exploration

INTRODUCTION

A mineral resource is a concentration or occurrence of material that has economic value on or above the earth's crust, with a certain shape, quality, and quantity that has reasonable prospects for eventually being economically extracted (Sukandarrumidi, 2007). Nickel laterite is one type of natural resource that cannot be renewed, this is because the formation of nickel laterite requires a process that takes millions of years (Thamsi, Jafar, and Fauzie 2021a; Yogi Pranata et al. 2017).

Some of the factors that form nickel include rock of origin, climate, chemical reagents and vegetation, structure, topography, and time (Waheed, 2002; Hardyanto 2016; Hasria et al. 2021; Yogi Pranata et al. 2017). The need for high-grade mills causes ore, ore itself is a valuable mineral that is sought and then extracted in mining activities with the hope (although not always achieved) to gain benefits for miners and for the community (Taylor, 1986). Low-grade ore will lose the selling price and will be categorized as waste because it cannot be processed at the factory (Habibie et al., 2019). This condition certainly requires proper handling so that low-grade ore is not wasted and the volume of reserves increases (Azizi et al., 2019; Ilyas et al., 2022; Sufriadin et al., 2012; Thamsi, 2017).

One of the unresolved problems in Region X is the determination of the cut-off grade. Region X itself has low levels or can be said to be incompatible with the products targeted by the company. The product targeted by the company is an ore which has a Ni content of 1.8%. Cut off grade or limit grade where the grade below has metal or mineral content in rock that does not meet economic requirements (Heriyawan et al, 2005 (Anshariah, 2016; Syah et al., 2018; Thamsi et al., 2021).

Copyright © 2022, Journal of Geology & Exploration, Page: 2

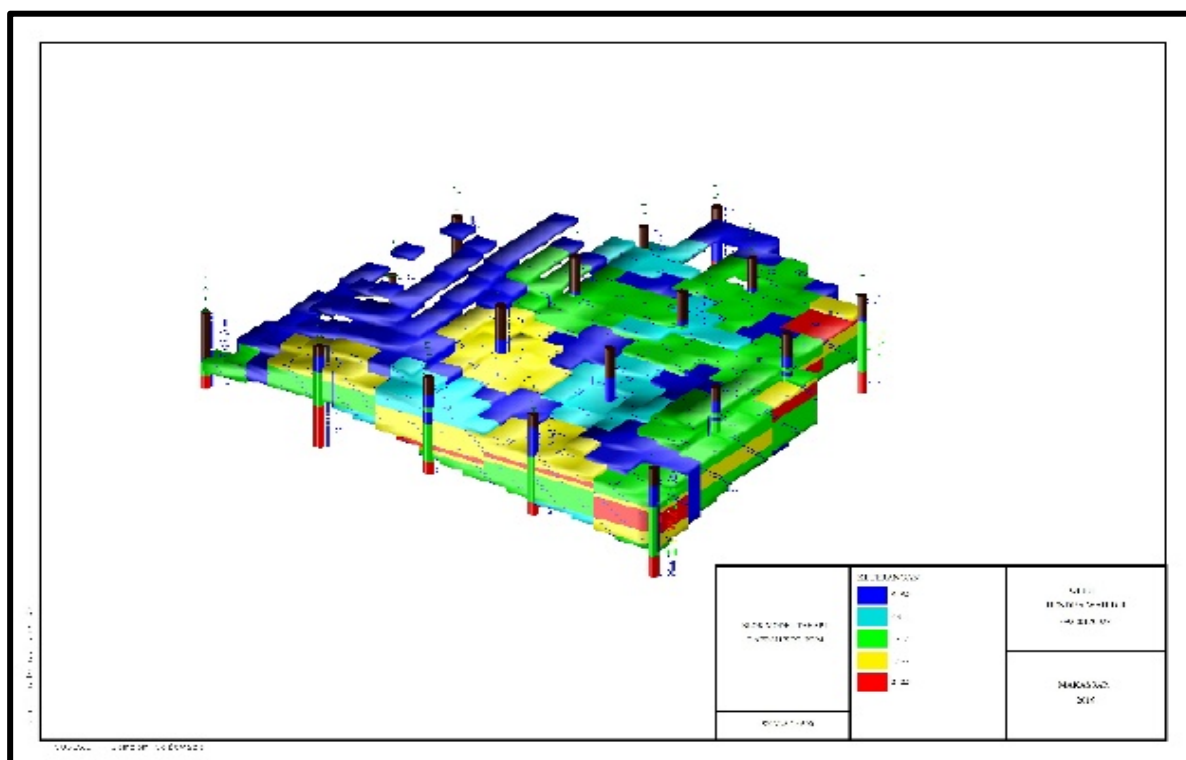


Figure 2. Ore model stage 1

In the first stage of the ore model block, nickel with levels above 1.5% dominates the area. In the first stage, the resulting ore tones are 89,937 tons, and the details can be seen in the following table.

Table 1. Details of ore at stage 1

If seen briefly in Figure 2, the ore with high levels is not very visible on the surface; this is

No.	Ni Class (%)		Volume (BCM)	Tonnage (tons)	Average Ni Content (%)
	From	To			
1	1	1.19	20375	29544	1.1
2	1.2	1.39	20725	30051	1.3
3	1.4	1.59	8900	12905	1.5
4	1.6	1.79	4675	6779	1.7
5	1.8	1.99	4825	6996	1.9
6	2	2.19	1675	2429	2.1
7	2.2	2.39	850	1233	2.3
8	2.4	2.59	0	0	0
9	2.6	2.9	0	0	0

because, in the second area, there is only a little more with high levels, which can be seen clearly in table 2 below.

From the results of the classification of levels of Ni at the last stage, nickel with the highest grade that can be obtained is nickel with a grade of 2.6 – 2.8% but with a very small tonnage of 979 tons with an average grade of 2.7% (Figure 3 and Table 3).

After calculating the formula for determining the total volume, average grade, and tonnage in each classification, the results obtained from these calculations can be seen in Table 4

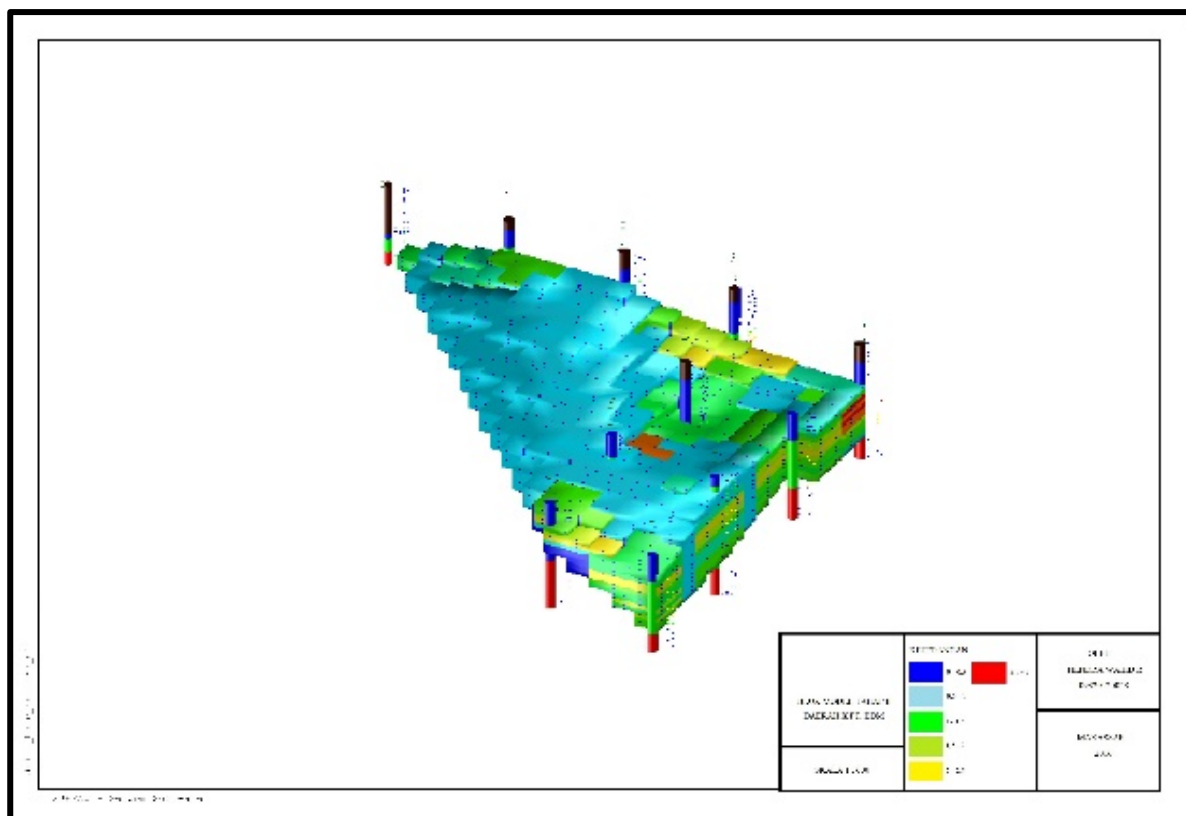


Figure 2. Model ore stage 2

Table 2. Details of ore at stage 2

No.	Ni Class (%)		Volume (BCM)	Tonnage (tons)	Average Ni Content (%)
	From	To			
1	1	1.19	41900	60755	1.1
2	1.2	1.39	39200	56840	1.3
3	1.4	1.59	25625	37175	1.5
4	1.6	1.79	21175	30704	1.7
5	1.8	1.99	12200	17691	1.9
6	2	2.19	8200	11890	2.1
7	2.2	2.39	5725	8301	2.3
8	2.4	2.59	1025	1486	2.4
9	2.6	2.9	700	1015	2.7

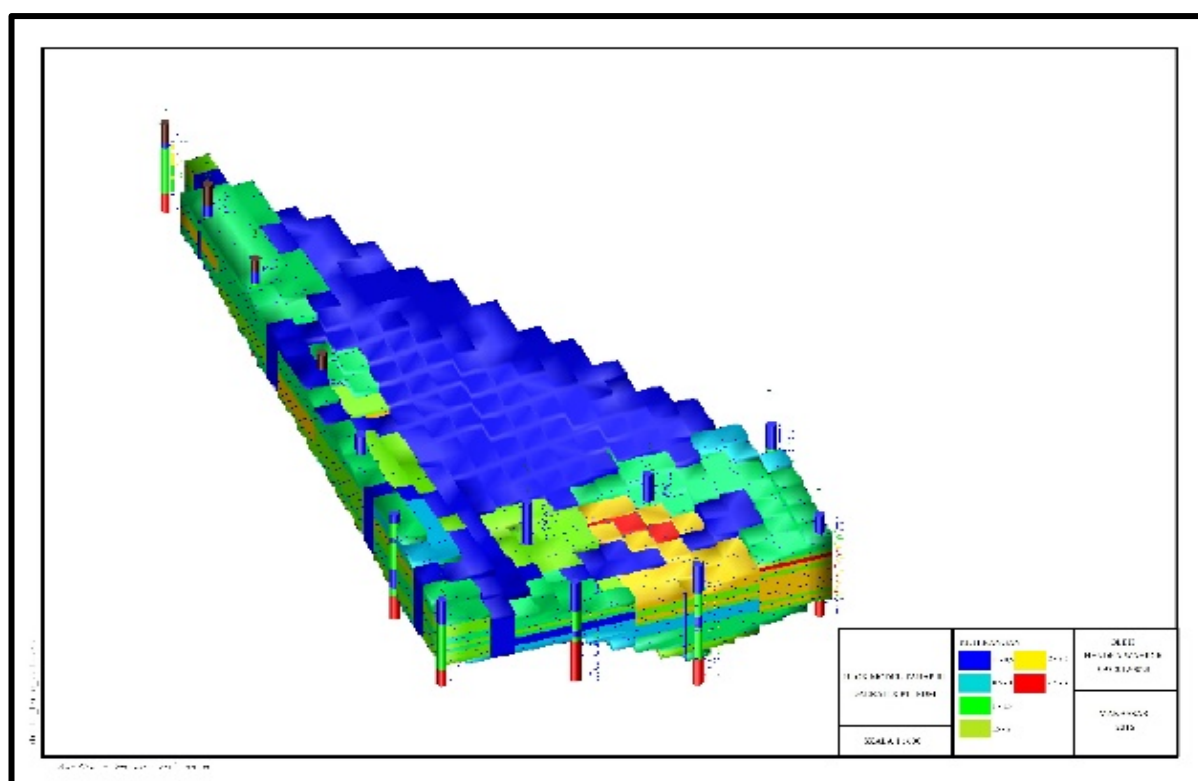


Figure 3. Model ore stage 3

Table 3. Details of ore at stage 3

No.	Ni Class (%)		Volume (BCM)	Tonnage (tons)	Average Ni Content (%)
	From	To			
1	1	1.19	41900	60755	1.1
2	1.2	1.39	39200	56840	1.3
3	1.4	1.59	25625	37175	1.5
4	1.6	1.79	21175	30704	1.7
5	1.8	1.99	12200	17691	1.9
6	2	2.19	8200	11890	2.1
7	2.2	2.39	5725	8301	2.3
8	2.4	2.59	1025	1486	2.4
9	2.6	2.9	700	1015	2.7

Table 4. Total area 1, 2, and 3

No.	Ni Class (%)		Volume (BCM)	Tonnage (tons)	Average Ni Content (%)
	From	To			
1	1	1.19	18325	26571	1.1
2	1.2	1.39	12775	18524	1.3
3	1.4	1.59	11800	17110	1.5
4	1.6	1.79	11525	16711	1.7
5	1.8	1.99	3700	5365	1.9
6	2	2.19	4650	6743	2.1
7	2.2	2.39	4925	7141	2.3
8	2.4	2.59	975	1414	2.4
9	2.6	2.9	675	979	2.7

Blending Stage Simulation

Blending is mixing materials of different quality and quantities to obtain an appropriate mixture. In the blending process, it must meet the specific ore needs of the processing plant (Musnajam, 2012). In the blending process, it must meet the specific ore needs of the processing plant, where the factory requires ore with a Ni content of 1.8%.

It can be done with the formula Average content to calculate the average level of blending ore results on the stockpile.

Formula:

$$\text{Average Ni Content} = \frac{(V_1 \times \text{Ni}) + \dots + (V_n \times \text{Ni}_n)}{V_{\text{total}}}$$

Table 5. Simulation of blending stages

Stage	Classification		Volume (BCM)	Tonnage (Ton)	Ni (%)	Average Ni Content (%)
	From	To				
1	2.2	2.39	5725	8301	2.3	2.07
	2.4	2.59	1025	1486	2.4	
	2.6	2.79	700	1015	2.7	
	2	2.19	8200	11890	2.1	
	1.8	1.99	12200	17691	1.9	
	Total		27850	40383		
2	1.8	2.39	27850	40383	2.07	1.9
	1.6	1.79	21175	30704	1.7	
	Total		49025	71086		

CONCLUSION

From the results of the discussion described previously, it can be concluded that the cut-off grade used in area X is 1.6, with a total tonnage of 71,086 tons by blending ore to optimize nickel with low grades.

From the above calculation results, the blending carried out in the first stage by mixing all ores with a high content of 1.8% and above resulted in a relatively high average content of 2.07%. This means that the lower grade can still be mined, so it is necessary to re-mix it with nickel with a lower grade classification. The results of the calculation of the average grade are possible to be used as a benchmark for determining the cut-off grade, namely the Ni classification, which has a grade above 1.6% with an average grade of 1.9%, and the total tonnage of ore that can be mined is 71.086 tons. This is proven in the calculation of the third stage of the simulation.

ACKNOWLEDGMENT

The authors would like to thank the supervisors in the research activities, Mr. Andika Purnama and Muhammad Haerulin, and all Mine Plan Engineering Division staff of PT. Bintang Delapan Mineral for providing opportunities, assistance, facilities, and guidance during research activities.

REFERENCE

- Anshariah, A. (2016). Estimasi Sumberdaya Nikel Laterit Dengan Metode In Verse Distance Weight Pada Kabupaten Konawe Utara Provinsi Sulawesi Tenggara. *Jurnal Geomine*, 4(1). doi: 10.33536/jg.v4i1.36
- Azizi, M. A., Hakim, R. N., & Nugraha, A. D. (2019). Optimalisasi Geometri Lereng Tambang Nikel Menggunakan Metode Probabilistik Pada Hill Pit 05 PT Vale Indonesia Tbk, Sorowako, Kabupaten Luwu Timur, Provinsi Sulawesi Selatan. *Jurnal Geomine*, 7(2), 92–100. doi: 10.33536/jg.v7i2.344
- Habibie, A., Widodo, S., Alim, M. N., Umar, E. P., Lantara, D., Nurwaskito, A., & Thamsi, A. B. (2019). Analisis Losses pada Pemindahan Material LGSO di Front Penambangan. *Jurnal Geomine*, 7(3), 212–218. doi: 10.33536/jg.v7i3.295



- Hardyanto, H. (2016). PEMODELAN ENDAPAN NIKEL LATERIT, KABUPATEN MOROWALI, PROVINSI SULAWESI TENGAH. *Jurnal Geomine*, 2(1). doi: 10.33536/jg.v2i1.29
- Hasria, H., Asfar, S., Ngkoimani, L. O., Okto, A., Jaya, R. I. M. C., & Sepdiansar, R. (2021). Pengaruh Geomorfologi Terhadap Pola Distribusi Unsur Nikel Dan Besi Pada Endapan Nikel Laterit Di Kabupaten Buton Tengah-Sulawesi Tenggara. *Jurnal GEOSAPTA*, 7(2), 103–114. doi: 10.20527/JG.V7I2.10716
- Heriyawan M.N., Notosiswoyo, S., Syafrizal, L dan Widayat, AH. (2005). “*Metode Perhitungan Cadangan TE-3231*”. Bandung. Departemen Teknik Pertambangan Fakultas Teknologi Mineral dan Kebumihan Institut Teknologi Bandung.
- Ilyas, A., Pasolo, A. R., & Widodo, S. (2022). Analisis Karakteristik Mineralogi dan Geokimia Berdasarkan Zona Profil Endapan Nikel Laterit (Studi Kasus: Blok X PT Ang and Fang Brother, Site Lalampu, Kecamatan Bahodopi, Kabupaten Morowali, Provinsi Sulawesi Tengah. *Jurnal Geomine*, 10(1), 01–12. doi: 10.33536/jg.v10i4.1165
- Musnajam. (2012). “*Optimalisasi Pemanfaatan Bijih Nikel Kadar Rendah Dengan Metode Blending Di PT. Antam Tbk. Ubpn Sultra*”. Kolaka. Jurnal Teknologi Technoscientia.
- Sufriadin, A., I., S., P., I., W. W., I., N., A., I., A., I. M., & Kaharuddin. (2012). Thermal and Infrared Studies of Garnierite from the Soroako Nickeliferous Laterite Deposit , Sulawesi , Indonesia Analisis Termal dan Inframerah Garnierit dari Endapan Laterit Nikel. *Indonesian Journal on Geoscience*, 7(2), 77–85. Retrieved from <http://ijog.bgl.esdm.go.id>
- Sukandarrumidi. (2007). “*Geologi Mineral Logam*”. Jogjakarta. Gajah Mada University Press.
- Taylor, H. K. 1986. “*On ore Reserve Estimation, Methods, Models, and Reality*”. *Can. Inst. Min Metal. Montreal. May*.
- Syah, M., Widodo, S., & Asmiani, N. (2018). Analisis Kelayakan Ekonomis Untuk Penentuan Pengadaan Alat Angkut Dan Alat Muat Pada Kegiatan Penambangan Nikel Sulawesi Tenggara. *Jurnal Geomine*, 6(2). doi: 10.33536/jg.v6i2.213
- Thamsi, A. B. (2017). Estimasi Cadangan Terukur Endapan Nikel Laterit Cog 2,0% Menggunakan Metode Inverse Distance Pada Pt. Teknik Alum Service, Blok X. *Jurnal Geomine*, 4(3), 128–130. doi: 10.33536/jg.v4i3.77
- Thamsi, A. B., Jafar, N., & Fauzie, A. (2021). Analisis Pengaruh Morfologi Pada Pembentukan Nikel Laterit PT Prima Sentosa Alam Lestari Kabupaten Morowali Provinsi Sulawesi Tengah. *Jurnal GEOSAPTA*, 7(2), 75. doi: 10.20527/jg.v7i2.9114
- Yogi Pranata, R., Djamaluddin, D., Asmiani, N., & Thamsi, A. B. (2017). Analisis Perbandingan Kadar Nikel Berdasarkan Perencanaan terhadap Realisasi Penambangan. *Jurnal Geomine*, 5(3), 143–146. doi: 10.33536/jg.v5i3.146
- Waheed, A. (2002). “*Nickel Laterites – A short course on the chemistry, mineralogy and formation of nickel laterites*”. Indonesia (unpublished). PT.INCO.