EXTRACTION OF NICKEL FROM A MIXED NICKEL-COBALT HYDROXIDE PRECIPITATE

KARIŞIK BİR NİKEL-KOBALT HİDROKSİT ÇÖKELEĞİNDEN NİKEL KAZANIMI

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ÖZ

The extraction of nickel from a mixed nickel-cobalt hydroxide precipitate (MHP) was conceived and experimentally explored. The effect of sulphuric acid concentration, leaching time, leaching temperature and stirring speed on nickel extraction was investigated. After 60 min leaching time, more than 90% of nickel and approximately 80% of cobalt were taken into the leach solution under the following conditions: 0.75M sulphuric acid concentration, 30°C leaching temperature and 400 rpm stirring speed. pH of the leach slurry was determined as 1.23 under the conditions explored. It was observed that there is no beneficial effect of leaching temperature and stirring speed on the extraction of nickel from the MHP sample. The XRD pattern of the fresh MHP sample revealed the data and peak of the sample widening which is indicative of amorphous solid structure.

ABSTRACT

Karışık bir nikel-kobalt hidroksit çökeleğinden (MHP) nikel kazanımı tasarılanmıştır ve deneyisel olarak araştırılmıştır. Nikel kazanımı için sülfürik asit miktarı, liç süresi, liç sıcaklığı ve karıştırma hızının etkileri incelenmiştir. 0,75M sülfürik asit miktarında, 30°C liç sıcaklığında ve 400 rpm karıştırma hızı şartlarında, 60 dakika liç ilesminden sonra %90 dan daha fazla nikel ve yaklaşık %80 kobalt çözüllüye alınmıştır. Araştırılan koşullar altında kati-çözelti karışımdaki pH değeri 1,23 olarak belirlenmiştir. Liç sıcaklığının ve karıştırma hızının MHP numunesindeki nikel kazanımı faydali etkilerinin olmadığı gözlemlemiştir. Yeni MHP numunesi üzerinde yapılan XRD incelemeleri sonrasında elde edilen sonuç ve piklerin oldukça geniş olması numunenin amorfmsu kati yapısı olduğu ortaya çıkartmıştır.

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INTRODUCTION

Lateritic nickel ore is an important natural mineral resource that contains important amount of nickel and its primary by-product, cobalt. These metals are extensively used in modern metallurgical applications such as stainless steel, nickel coating, non-ferrous alloys, cobalt coating on copper surface, batteries and mobile phones. Due to the decrease in high-grade ore deposits and an increase in demand from metallurgical industry for strategic applications, researchers have concentrated on low-grade lateritic nickel ores.

In the last two decades, several new hydrometallurgical technologies have been proposed for nickel and cobalt extractions from laterite nickel ore such as sulphating atmospheric leaching (SAL), starved acid leaching technology (SALT), bioleaching of oxide ores, and Direct Nickel process (DNi) (Verbaan et al., 2007; Harris et al., 2004; McDonald and Whittington, 2008; Dreisinger and Clucas, 2013; Valix et al., 2001; Kyle, 2010; Brock and McCarthy, 2014). The main novelty of SAL process is that the concentrated sulfuric acid is added directly to lateritic nickel ore in a pug mill. Relatively low levels of sulfuric acid were used to dissolve nickel and cobalt from laterite ores using the SALT method, in conjunction with relatively low levels of iron dissolution. As an alternative to hydrometallurgical process, bio-hydrometallurgy has gained increasing attention because of the potential lower cost, lower energy demand, and safer operation compared to the conventional hydrometallurgical processes. In this process, the extraction of nickel and cobalt from laterite nickel ore occurs with a range of fungal strains (i.e. Aspergillus niger, Penicillum funiculosum) that produce organic acids such as citric acid, oxalic acid, and acetic acid which enable the complexation of the metals (Tang and Valix, 2006). DNi process has been demonstrated at demonstration scale test plant (Brock and McCarthy, 2014). In this process, nitric acid is used instead of sulfuric acid to dissolve nickel within a few hours and the residual acid is then recycled. The main advantage of the process is that the process treats both limonite and saprolite ores and it is estimated to have less than half the capital and operating costs of HPAL or ferronickel processes.

There are three leaching methods currently applied for the hydrometallurgical extraction of nickel and cobalt from lateritic nickel ores, namely high pressure acid leaching (HPAL), atmospheric acid leaching (AL) and heap leaching (HL). Each process has its own advantages and disadvantages. The main advantages of HPAL are that the process requires much lower acid consumption, higher nickel and cobalt recoveries and faster dissolution kinetics than the other two methods. The main advantages of AL are that it requires lower capital and energy costs than HPAL but able to provide comparable nickel and cobalt recoveries. Meanwhile, HL requires the least capital and energy costs among the three processes and produces clear leachate solution for downstream processing circumventing the need of a solid-liquid separation unit. This process, however, provides the least metals recoveries among the three, requires long leaching duration, inventory and cycle management (Kyle, 2010).

Recently, nickel and cobalt are dissolved by direct sulphuric acid leaching method from lateritic nickel ore in industrial scale. Nickel and cobalt are extracted from pregnant leach solution (PLS) by one of three methods. These methods are divided into three main categories depending on the downstream route: (i) mixed sulphide precipitation (MSP), (ii) mixed hydroxide precipitation (MHP) and (iii) direct solvent extraction (DSX) (Motteram et al., 1996; Manson et al., 1997; Mihaylov et al., 2000; Donegan, 2006). The MSP route has been in commercial operation for more than 55 years and is now the most used technique to recover the nickel and cobalt as a filter cake of mixed nickel and cobalt sulphides separating them from manganese, magnesium and calcium. It has been used at Moa Bay (Cuba), Murrin Murrin (Australia), Coral Bay (Philippines), Ambatovy (Madagascar) and Taganito (Philippines). The MHP route has been used at Cawse (Australia), Ravenstorphe (Australia) and Ramu
Papua New Guinea). This process is simpler to operate than the MSP process and is unsuitable for treating feed liquor with high manganese content. The DSX route is the newest addition among the three and has now been used in two commercial operations: Bulong (Australia) and Goro (New Caledonia). This process circumvents the need of intermediate precipitation of the nickel and cobalt and therefore, this approach offers a potential economic advantage over the two intermediate precipitation processes.

Nickel and cobalt are separated from each other and extracted from the resulting solutions by hydrogen reduction, electro-winning (EW) or precipitation. MSP route excludes more of the common impurities whereas MHP is increasingly popular because it omits the costly H₂S precipitation step and its product is easily soluble in ammonia or dilute sulphuric acid with a potential for the application of SX/EW (Taylor, 1995).

It was noted that DSX had a high level of impurities like magnesium, calcium and iron (II) as well as considerable amount of gypsum formation and crud build-up formations in solvent extraction settlers (Donegan, 2006; Nofal et al., 2001; O’Callaghan, 2003). Due to high operating cost and technical risk of DSX, MHP route has been developed for refining with a simple and effective nickel and cobalt separation step (Mason and Hawker, 1998).

Hydrometallurgical route of nickel laterite ore is complex because of impurities such as aluminium, chromium, iron, calcium, magnesium and manganese. Cobalt, especially, is a value metal which is always present in lateritic nickel ores. After leaching of lateritic nickel ore, aluminium, chromium and iron are precipitated by CaCO₃, CaO or MgO, leaving nickel, cobalt, manganese, calcium and magnesium in solution. Further, sulphide or hydroxide precipitation is conducted to achieve manganese, calcium, magnesium separation from nickel and cobalt. Hydroxide precipitation; however, is less selective towards manganese (Harvey et al., 2011).

High pressure acid leaching (HPAL) technique is currently used by Meta Nickel, Gordes/Turkey to extract nickel and cobalt from lateritic nickel ore. PLS is fed to first partial neutralisation unit to remove impurities such as iron, aluminium, chromium by increasing pH of the leach solution up to ~3.3. The neutralised leach solution is send to second neutralisation unit to achieve complete rejection of the impurities by increasing pH of the leach solution from ~3.3 to ~5.5 and then the leach solution is precipitated by using MgO to obtain MHP product. The MHP product depending on PLS consists of ~34-38%Ni and ~1.8-2.1%Co along with other impurities such as manganese, magnesium and calcium. In the present study, the main objective was to investigate the extraction of nickel from the MHP product using direct sulphuric acid leaching under atmospheric conditions to understand the effect of sulphuric acid concentration, leaching temperature, stirring speed and leaching time.

1. MATERIALS AND METHOD

1.1. MHP Sample

The fresh MHP sample was kindly provided by Meta Nickel Plant in Gordes/Turkey. The supplied sample was crushed with a jaw crusher and reduced to less than 425 µm using a laboratory ball mill. The fresh MHP sample was given in Figure 1. The particle size distribution of the sample was measured using Master Sizer 2000 (Malvern). The elemental composition of the sample was analysed using X-ray fluorescence (XRF) spectrometer (Zetium, PANalytical), while its mineralogical composition was examined using X-ray diffraction (XRD) (Empyrean, PANalytical) with CuKα₁ (wavelength 1.5406 Å) operating at 45 kV and 40 mA. The XRD 2θ diffraction angle was calibrated using 2θ/d-spacing calibration standard. Subsequently, minerals were identified using the ICCD PDF-4 Plus/Minerals software. The wt.% of nickel was determined on a Thermo Scientific 3300 atomic absorption spectrophotometer (AAS) after digestion of the fresh MHP sample in hot aqua regia (3:1, HCl:HNO₃).
1.2. Experimental Procedure

The leaching tests were performed in an approximately 600 cm³ glass reactor covered with aluminium foil to prevent evaporation. The leach slurry was mixed by an overhead mechanical stirrer (MTOPS-MS3040D) with a three-blade stainless steel impeller (45 mm diameter) at 400 rpm. The reactor and slurry were maintained at the required temperature by a hotplate heater. For each experiment, 50 g ore were transferred into the reactor before adding 500 cm³ of sulphuric acid at a pre-determined concentration and then heated to the desired temperature. At selected time intervals, a fix amount of slurry (10 cm³) was withdrawn, and the pH of the slurry was measured by pH meter (Hach, HQ40d) equipped with IntelliCAL PHC 28101 probe, and lastly the slurry was centrifuged at 7500 rpm. Supernatant liquor was diluted and analysed by AAS. In the leaching tests, analytical grade of sulphuric acid (95-97% purity, Merck) was used in acid preparation. In all leaching tests, solid-to-liquid ratio was maintained at 1/10 (w/v). The leaching tests were duplicated to assess the reproducibility of the test results. Extraction of nickel was calculated according to the following equation:

$$\text{Extraction} \ (%)=\left[\frac{\text{Metal in the leach liquor by AAS, %}}{\text{Metal in the fresh MHP sample by AAS, %}}\right] \times 100$$ (1)

2. RESULTS AND DISCUSSION

2.1. Materials Characterization

The particle size distribution of the sample is shown in Figure 2. The results showed that 90% of the sample is less than 43.1 µm. The elemental composition of the sample was given in Table 1. The elemental concentrations of nickel and cobalt were determined as 34.5 and 1.9%, respectively by AAS. Figure 3 shows a XRD pattern of the MHP sample.

Table 1. Elemental composition of the MHP by XRF

<table>
<thead>
<tr>
<th>Elements (MHP)</th>
<th>wt.%</th>
</tr>
</thead>
<tbody>
<tr>
<td>Mn</td>
<td>5.63</td>
</tr>
<tr>
<td>Fe</td>
<td>0.56</td>
</tr>
<tr>
<td>Cr</td>
<td>0.04</td>
</tr>
<tr>
<td>Sc</td>
<td>0.02</td>
</tr>
<tr>
<td>Zn</td>
<td>0.45</td>
</tr>
<tr>
<td>Cu</td>
<td>0.22</td>
</tr>
<tr>
<td>Co</td>
<td>2.1</td>
</tr>
<tr>
<td>Ni</td>
<td>37.7</td>
</tr>
<tr>
<td>Mg</td>
<td>2.42</td>
</tr>
<tr>
<td>Al</td>
<td>0.9</td>
</tr>
</tbody>
</table>

The XRD results revealed the data and peak of the sample widening which is indicative of amorphous solid structure. It is known that determination of amorphous structure of solid sample is difficult. The characteristic peaks of $\alpha$-Ni(OH)$_2 \cdot 0.75$H$_2$O
(card No: 00-038-0715-Jamborite) nickel hydroxide obtained from sulphate solution achieved at 2-Theta= 34 and 59. The peaks at 2-Theta= 12 and 23 were also determined as characteristics of α-Ni(OH)2·0.75H2O, which is consistent with findings of Rajamathi et al., 1997, Harvey et al., 2011.

Figure 3. XRD pattern of the MHP sample

2.2. Effect of Sulphuric Acid Concentration on Extraction

A series of leaching tests was carried out at various sulphuric acid concentrations (0.5, 0.75 and 1.0M). The leaching time, temperature, stirring speed and solid-to-liquid ratio were fixed at 3h, 70°C, 400 rpm and 1/10, respectively. Equation 2 shows the precipitation of metals from aqueous solution as hydroxides. M is the metal cation and n is the cation charge. According to Equation 3, nickel hydroxide precipitate can dissolve in aqueous sulphuric acid solution.

\[
M^{n+} + nH_2O = M(OH)_n + nH^+ \tag{2}
\]

\[
Ni(OH)_2(s) + H_2SO_4 (aq) \rightarrow NiSO_4 (aq) + H_2O(l) \tag{3}
\]

Figure 4 shows the effect of sulphuric acid concentration on the nickel extraction from the MHP sample at 70°C. The extraction rate of the nickel significantly increased with increasing concentration of sulphuric acid. The extraction of nickel increased from 83.2% to 96.2% when the acid concentration increased from 0.5 M to 0.75 M at 70 °C for 3h and then relatively maintained constant. It may be due to acid consuming alkaline phases such as magnesium and calcium hydroxide in the product. Figure 5 shows the slurry pH changes. The pH of the slurry decreased from 4.9 to 1.0 when the acid concentration increased. Thus, nickel extraction rate increased depending on ionic activity of the solution.

2.3. Effect of Leaching Time on Extraction

Leaching experiments was conducted with sulphuric acid concentration of 0.75M and temperature of 70°C as a function of time with a 1/10 (w/v) solid-to-liquid ratio and a stirring speed of 400 rpm. Figure 6 shows the effect of leaching time on the nickel extraction from the MHP sample. The extraction of nickel increased with increasing leaching time up to 60 min and then slightly increased until 180 min. As can be seen from Figure 6, the extraction of nickel is achieved within 60 min. Therefore, the leaching time should be selected as 60 min for the extraction of nickel from the MHP sample. Figure 7 shows the slurry pH changes during the leaching process. The pH of the slurry slightly increased within the first 30 min and then stayed relatively constant. This means that the ionic activity of the solution remained constant when the leaching time extended from 60 min to 180 min.

Figure 4. Effect of sulphuric acid concentration on extraction (Conditions: Temperature of 70°C, leaching time of 180 min, solid-to-liquid ratio of 1/10, stirring speed 400 rpm)
2.4. Effect of Stirring Speed on Extraction

The effect of the stirring speed on the nickel extraction was evaluated at 500-600 rpm and 70°C with 0.75 M sulphuric acid as a function of leaching time (Figure 8). The solid-to-liquid ratio was held constant at 1/10 (w/v). Nickel extraction slightly increased with increasing leaching time. As can be understood from Figure 8, there was no beneficial effect of stirring speed on the extraction of nickel. Therefore, the stirring speed can be held at 400 rpm to achieve homogenous mixing of the leach slurry.

2.5. Effect of Temperature on Extraction

The leaching curves of nickel and cobalt at different temperatures are shown in Figure 9. The following conditions were performed in the effect of temperature studies: a 1/10 (w/v) solid-to-liquid ratio, 0.75 M sulphuric acid, 180 min leaching time and a constant stirring speed of 400 rpm. The extraction of nickel slightly increased with increasing leaching temperature whereas the extraction of cobalt remained relatively constant. As can be seen from Figure 9, there is no benefit gained in the leaching of nickel and cobalt by increasing leaching temperature. Therefore, the extraction of nickel can be achieved at 30°C. These results are in agreement with the finding of Harvey et al., 2011.

Figure 6. Effect of leaching time on extraction (Conditions: Sulphuric acid concentration of 0.75M, temperature of 70°C, solid-to-liquid ratio of 1/10, stirring speed of 400 rpm)

Figure 7. pH changes of the leach slurry

Figure 8. Effect of stirring speed on extraction (Conditions: Sulphuric acid concentration of 0.75M, temperature of 70°C, solid-to-liquid ratio of 1/10)

Figure 5. pH changes of the leach slurry
Nickel was extracted from a mixed nickel-cobalt hydroxide precipitate with excellent extraction under atmospheric pressure acidic leaching conditions. The effects of four experimental variables were evaluated. After leaching experiments, the maximum nickel extraction in this study was achieved under the following conditions: 0.75 M sulphuric acid, 30°C, a 1/10 solid-to-liquid ratio (w/v), a leaching time of 60 min, and a stirring speed of 400 rpm. Under these conditions, more than 90% of the nickel and approximately 80% of cobalt were taken into the leach solution. The extraction of nickel slightly increased after 30 min leaching time and then remained constant while the extraction of cobalt was relatively stable. It was found that there is no influential effect of leaching temperature and stirring speed on the extraction of nickel. Effective nickel extraction could be achieved using 30°C leaching temperature from the mixed nickel-cobalt hydroxide product under the conditions explored. The XRD result of the MHP sample demonstrated that sample has an amorphous solid structure.

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