Introduction

In recent years, the demand for copper and nickel has risen world-wide due to their use in printed circuit boards, batteries, application in marine alloys (due to excellent resistance to corrosion by seawater), and a wide range applications in day-to-day life (Doebrich, 2009). Increasing demand for these metals is leading to rapid depletion of high-grade resources. Alternative sources, such as e-wastes, lithium ion batteries, copper converter slags, polymetallic nodules etc., are needed to meet future demand. The copper-nickel system was chosen for this investigation because some slags (copper converter slag, anode slag of Hindustan Copper, and by-products of UCIL Jaduguda) contain high amounts of both copper and nickel. Hydrometallurgical routes play a vital role in the processing of these secondary sources. Solvent extraction is a well-established technique for the removal and separation of various metal ions. About 30% of the world’s copper is produced by acid leaching, solvent extraction (SX), and electrowinning (Cox, 2004). Hydroxyoximes are well-known extractants which are widely used for copper extraction from dilute acidic sulphate solutions (Ritcey and Ashbrook, 1979; Szymanowski, 1993; Szymanowski, 1990).

Only two classes of extractants, i.e. ketoximes and aldoximes, have gained commercial acceptance as reagents for solvent extraction from acidic leach solutions. Oxime mixtures have been shown to be advantageous over the individual group ketoximes and aldoximes (Kordosky et al., 1985). Rodriguez et al. (1997) studied the extraction of several metal ions using LIX® 984, which is a 1:1 volume blend of LIX® 860 and LIX® 84 in n-heptane. They reported the equilibrium constant values for the extraction reactions, and proposed the species extracted into the organic phase. Fouad (2007) studied the extraction equilibria of copper(II) with Cyanex 301, LIX® 984N, and mixtures of these two reagents. The enthalpy change (ΔH) of the individual extractants and their mixtures was determined and the endothermic process elucidated. Kul and Cetinkaya (2009) developed a complete hydrometallurgical process on a laboratory scale for recovering copper from copper electroplating second rinse bath liquor containing 2.5 g/L copper using LIX® 984N-C in kerosene. Recovery of copper from waste printed circuit boards (PCBs) by nitric acid leaching and extraction using LIX® 984N was reported by Long Le et al. (2011). They first studied the co-extraction of other metals (Pb, Zn, Fe, and Ni) from a diluted leach liquor with 10% LIX® 984N, and found that the co-extraction of these metals was negligible up to pH 1.9, except for iron (6.4%). Based on these results, they used 50% LIX® 984N-C in kerosene. Further studies are needed to investigate the co-extraction of other metals from diluted leach liquors using LIX® 984N-C.
Liquid-liquid extraction and separation of copper(II) and nickel(II) using LIX® 984N

984N to recover all the copper from the actual leach liquor of composition 42.11 g/L Cu, 2.12 g/L Fe, 4.02 g/L Pb, 1.58 g/L Zn, and 0.48 g/L Ni at an A:O ratio of 1:1.5 in three stages and at pH 1.5. Sulphuric acid (H2SO4 -360 g/L) was used for stripping. Qing-ming et al. (2008) studied the separation of copper and iron from dump bioleaching solution from Dexing Copper Mine using LIX® 984N. Kinoshita et al. (2003) reported the separation of copper and nickel from nitrate media containing 11 180 mg/L Cu and 1160 mg/L Ni using 200 g/L LIX® 984. Extraction and separation of copper and nickel by LIX® 984N from ammoniacal medium was investigated by Sridhar et al. (2009). They found that both the metals co-extracted over the pH range 7 to 9. A selective stripping method was used to separate copper and nickel.

The use of LIX® 984N has received attention for extracting copper from copper-rich leach liquors. There are few reports available regarding equilibrium studies of the copper-LIX® 984N system (Aminian and Bazin, 2000) and the separation of copper and nickel from sulphate medium (Ochromowicz and Chmielewski, 2013). The present study investigates the extraction equilibrium in the copper-LIX® 984N system, the species extracted into the organic phase, and evaluates the thermodynamic parameters in addition to diluent and salt effects. McCabe-Thiele plots and batch countercurrent extractions were also been carried out to study the separation and recovery of copper from a synthetic copper and nickel solution.

Experimental

Solutions and reagents
Stock solutions of copper(II) and nickel(II) (1M each) were prepared by dissolving the sulphate salts in double-distilled water. LIX® 984N (a mixture of 5-nonylhexylidoxime and 2-hydroxy-5-nonylacetophenone oxime) was supplied by Cognis Inc. and was used without further purification. Distilled kerosene was used as diluent for the organic phase. All other reagents used were of analytical reagent grade.

Experimental methods

Solute extraction
The metal-bearing solution was equilibrated with an equal volume of LIX® 984N for 5 minutes in a separating funnel. The pH of the aqueous solution before extraction was adjusted by adding dilute H2SO4 or NaOH solution. After phase separation the aqueous phase was collected and the equilibrium pH was measured. The metal content in the aqueous phase was determined by the thiosulphate method using starch as the indicator (Bassett et al., 1984). When both copper and nickel were present, the aqueous phase was analysed by atomic absorption spectrophotometry (AAS) using an ELICO-type instrument. The concentration of the metal ion in the organic phase was calculated by difference in concentration before and after extraction. When required, the organic phase metal concentrations were determined after filtration through 1PS paper separating paper and stripping with 20% H2SO4, followed by analysis with AAS. All the extraction and stripping experiments were carried out at 30±1°C, except for the temperature variation study.

Countercurrent SX process
A two-step countercurrent extraction process was simulated by batch experiments using a synthetic solution containing copper (0.35 g/L) and nickel (0.58 g/L) up to five cycles at an O:A ratio of 3:4. In each step, 80 mL of aqueous phase and 60 mL of organic phase (20% (v/v) LIX® 984N) were mixed in a separating funnel for 5 minutes. A schematic representation of countercurrent steps is given in Figure 1. The raffinates and the loaded organic phases (after stripping with H2SO4) were analysed by AAS. Similar steps were carried out for the batch countercurrent stripping process, which involves three stages at an O:A ratio of 3:1.

Results and discussion

Extraction of copper using LIX 984N

Effect of equilibrium pH
Experiments were carried out at room temperature to study the effect of equilibrium pH on the extraction of 0.1M Cu(II) using 15% (v/v) LIX® 984N in kerosene. The results are presented in Figure 2 as percentage of copper extraction versus equilibrium pH. As expected, the percentage extraction increased with increasing equilibrium pH; however, it remained practically constant at equilibrium after sometime. At an equilibrium pH of 0.53, the percentage copper extraction into the organic phase was 78%, whereas 87% copper was extracted at equilibrium pH 0.61 and 88% at equilibrium pH 0.71. This behaviour was also reported by Asghari et al. (2009) and Rodriguez et al. (1997), who found that copper extraction increased with increasing equilibrium pH to a certain value, after which it was independent of pH variation. In this work, the maximum extraction of copper was 88% (equilibrium pH range 0.63 to 0.71) with 15% LIX®984N. These extraction and equilibrium pH values are low compared to the values reported by Asghari et al. (2009). One reason may be the higher initial copper concentration –

Figure 1—Schematic diagram of countercurrent extraction

Extraction and stripping isotherms
While keeping the total volume of the aqueous and the organic phases constant, the solutions at different O:A ratios were shaken for 5 minutes at 30±1°C. After the extraction process, the metal concentrations in the raffinate (R) were determined by AAS and the metal concentration in the loaded organic (LO) phase was calculated by mass balance. After the stripping process, the metal concentration in the spent organic (SO) was determined by AAS and the metal in the aqueous calculated by mass balance. McCabe-Thiele constructions were drawn for the extraction and stripping isotherms.

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6.35 g/L used in this study, compared with 2.5 g/L used by Asghari et al. Due to the availability of more copper ions, extraction is favoured and the release of more H⁺ ions results in a decreasing equilibrium pH value.

Effect of extractant concentration

The effect of extractant concentration on copper extraction was studied by contacting sulphate solutions containing 0.1M copper(II) at an initial pH value of 3.95 with LIX® 984N at concentrations from 5–25% (v/v). As shown in Figure 3, copper extraction increases from 2.8 g/L (44%) to 6.27 g/L (98.7%) with increase in extractant concentration from 5% to 25% (v/v) LIX® 984N.

Extraction mechanism

As LIX® 984N is a chelating extractant, it will form neutral complexes with copper(II) coordinating with the nitrogen atom of the oxime group and releasing two H⁺ ions into aqueous solution. The extraction equation can be written as:

\[ \text{Cu}^{2+} + n\text{[HR]}_{\text{org}} \rightleftharpoons \text{CuR}_n(\text{aq}) + 2\text{H}^+ (\text{aq}), \]  

where \( n \) is the solvation number of the extractant, LIX® 984N.

The extraction equilibrium constant can be represented as:

\[ K_{\text{ex}} = \frac{[\text{CuR}_n(\text{aq})][\text{H}^+]^{2n}}{[\text{Cu}^{2+}][n\text{[HR]}_{\text{org}}]} \]  

The distribution ratio \( (D) \) is the ratio of metal concentration in organic phase to the metal concentration in aqueous phase at reaction equilibrium. Substituting \( D \) in Equation [2] yields

\[ K_{\text{ex}} = \frac{D[H^+]^2}{[\text{HR}]_{\text{org}}} \]  

Therefore,

\[ \log D = \log K_{\text{ex}} + 2pH + n \log [\text{HR}]_{\text{org}} \]  

The concentration of copper chosen for the present study is 0.1M, and the results of the pH variation study led to the consideration of the equilibrium concentration of the organic phase (free extractant concentration remained after equilibration). This can be calculated as:

\[ [\text{HR}]_{\text{equ}} = [\text{HR}]_{\text{initial}} - n[H^+]_{\text{org}} \]  

The use of various concentrations of LIX 984 to determine the solvation number was reported by Rodriguez et al. (1997), Aminian and Bazin (2000), and Fouad (2009). The plot of \( \log D - \log [\text{HR}]_{\text{org}} \) versus equilibrium pH for different values of the number of extractant molecules \( n \) is shown in Figure 4. The line that has a slope of approximately 2.0 has been selected. The slope is equal to the number of H⁺ ions in Equation [1].

Based on the above results, the extraction equilibrium reaction is written as:

\[ \text{Cu}^{2+}(\text{aq}) + 2[\text{HR}]_{\text{org}} \rightleftharpoons \text{CuR}_2(\text{org}) + 2\text{H}^+ (\text{aq}) \]

Buketova (2009) analysed the IR spectrum and confirmed that the structure of this CuR₂ complex with LIX® 984N is non-polar. The intercept of the graph in Figure 4 is the logarithmic value of the extraction equilibrium constant. The equilibrium constant \( (K_{\text{ex}}) \) for the above equilibrium is therefore 2.90 and the Gibbs free energy of the extraction is -2.69 kJ mol⁻¹ (\( T = 303 \) K).

Effect of different media

Some metals are recovered via a hydrometallurgical route consisting of three unit operations – leaching, SX, and electrowinning. The SX process depends mainly on the nature of the leach liquor. In addition to leach liquors in a single medium, recovery of metals from mixed-medium leach liquors are also of importance nowadays (Sarangi et al., 2007; El-Hefny et al., 2010). Sarangi et al. studied the separation and recovery of iron(III), copper(II), and zinc(II) from mixed sulphate and chloride media, whereas El-Hefny et al. used sulphate/thiocyanate media for zinc and cobalt separation. Keeping this in mind, the extraction of 0.1M Cu(II) was carried out with 15% (v/v) LIX® 984N in the

Figure 2—Effect of equilibrium pH on copper extraction (0.1 M) using 15%(v/v) LIX® 984N in kerosene

Figure 3—Effect of extractant concentration on copper extraction (0.1M)
Liquid-liquid extraction and separation of copper(II) and nickel(II) using LIX® 984N

Presence of different salts such as sodium sulphate, sodium chloride, and sodium nitrate over the concentration range 0.1 to 1M. The results showed that the distribution coefficient of copper increased with increasing sodium sulphate concentration from 0.1M ($D = 9.08$) to 1 M ($D = 17.68$), but decreased with increase in sodium chloride concentration. In presence of 0.1M sodium nitrate, the distribution coefficient is more than the distribution coefficient value when sodium nitrate was absent, but it decreased constantly with further increase of NaNO₃ concentration (Table I). However, the decrease is more significant with sodium chloride than with sodium nitrate. The decrease in extraction may be due to a salting-out effect.

**Effect of diluent**

The choice of diluent is an important aspect of successful SX operation. A diluent reduces the viscosity of the extractants, but sometimes the nature of diluent affects the extraction process. To investigate this, the extraction of copper(II) with 15% (v/v) LIX® 984N was carried out using different diluents such as kerosene, benzene, xylene, toluene, carbon tetrachloride, and chloroform. The result showed that copper extraction depends on the dielectric constant value of the diluent – the higher the value of the dielectric constant, the lower the percentage extraction of copper (Table II). The reason for this is that with an increase in the dielectric constant of the diluent, the interaction between the diluent and the extractant is increased, thus decreasing the availability of the extractant for extraction of copper. The same trend was observed by Reazai and Nedjate (2003) and El-Nadi (2010) while studying the effect of diluents on the extraction of nickel and rare earths.

**Effect of temperature**

The extraction efficiency of certain extractants depends significantly on temperature, and therefore the extraction of 0.1M Cu(II) was studied at different temperatures (293–323K) with 15% (v/v) LIX® 984N in kerosene. It was observed that with increasing temperature the distribution ratio of copper extraction increased from 7.35 (299K) to 9.08 (323K). This indicates that the extraction is endothermic. The thermodynamic parameters such as enthalpy change ($\Delta H$) and entropy change ($\Delta S$) were calculated by plotting log $K_{eq}$ against $1000/T$ (Figure 5). From the plot, $\Delta H$ and $\Delta S$ were calculated and found to be 10.224 kJ mol⁻¹ and 4.05 J K⁻¹ respectively. Aminian and Bazin (2000) also reported that the extraction of copper with LIX® 984 was endothermic, with a $\Delta H$ of 5.88 kJ mol⁻¹ under the experimental conditions used (0.94 g/L Cu, pH = 2 and O: A = 0.1). Although in both cases the process was found to be endothermic, the difference in $\Delta H$ values may be due to the different experimental conditions.

**Table I**

Effect of different media on extraction of 0.1M Cu(II) with 15% (v/v) LIX® 984N in kerosene

<table>
<thead>
<tr>
<th>Distribution coefficient</th>
<th>$[\text{Salt}]$, M</th>
<th>Na₂SO₄</th>
<th>NaCl</th>
<th>NaNO₃</th>
</tr>
</thead>
<tbody>
<tr>
<td>0</td>
<td>7.47</td>
<td>7.47</td>
<td>7.47</td>
<td></td>
</tr>
<tr>
<td>0.1</td>
<td>9.08</td>
<td>6.74</td>
<td>9.08</td>
<td></td>
</tr>
<tr>
<td>0.2</td>
<td>10.98</td>
<td>6.41</td>
<td>9.08</td>
<td></td>
</tr>
<tr>
<td>0.3</td>
<td>13.27</td>
<td>6.15</td>
<td>8.62</td>
<td></td>
</tr>
<tr>
<td>0.6</td>
<td>14.88</td>
<td>5.90</td>
<td>8.10</td>
<td></td>
</tr>
<tr>
<td>0.8</td>
<td>16.21</td>
<td>5.29</td>
<td>7.70</td>
<td></td>
</tr>
<tr>
<td>1.0</td>
<td>17.68</td>
<td>4.89</td>
<td>7.36</td>
<td></td>
</tr>
</tbody>
</table>

**Table II**

Effect of diluent on extraction of 0.1M Cu(II) with 15% (v/v) LIX® 984N

<table>
<thead>
<tr>
<th>Diluents</th>
<th>Dielectric constant</th>
<th>$D$</th>
</tr>
</thead>
<tbody>
<tr>
<td>Chloroform</td>
<td>4.81</td>
<td>3.26</td>
</tr>
<tr>
<td>Xylene</td>
<td>2.57</td>
<td>4.25</td>
</tr>
<tr>
<td>Toluene</td>
<td>2.38</td>
<td>4.75</td>
</tr>
<tr>
<td>Benzene</td>
<td>2.27</td>
<td>5.29</td>
</tr>
<tr>
<td>Carbon tetrachloride</td>
<td>2.24</td>
<td>5.48</td>
</tr>
<tr>
<td>Kerosene</td>
<td>1.8</td>
<td>7.47</td>
</tr>
</tbody>
</table>

Figure 4—Plot of log $D$ - $n/2$ log ($[HR]$) vs equilibrium pH. Cu(II) (0.1M), 15% (v/v) LIX® 984N

Figure 5—Plot of log $K_{eq}$ vs 1000/T. Cu(II) (0.1M), 15% (v/v) LIX® 984N
Solvent extraction behaviour of copper in presence of nickel

To study the extraction behaviour of copper in presence of nickel, an aqueous solution containing 0.1M Cu(II) (6.35 g/L) and 0.01M Ni(II) (0.58 g/L) was chosen. Copper extraction was 98% with 25% (v/v) LIX® 984N, and 93% with 20% (v/v) LIX® 984N. In an industrial application the viscosity of the solvent should not be very high, therefore 20% (v/v) LIX® 984N in kerosene was selected to study copper-nickel extraction. Figure 6 represents the percentage extraction of Cu(II) and Ni(II) as a function of equilibrium pH of the aqueous solution. Extraction of copper increased from 61% to 91% with an increase in equilibrium pH from 0.36 to 0.91, while the extraction of nickel was within 6% to 8% over the equilibrium pH range 0.56 to 1.08. Kinoshita et al. (2003) reported the extraction of nickel to be 3–4 % around that range of equilibrium pH in the presence of copper. This result demonstrated that copper can easily be separated from nickel. The separation factor ($\beta = \frac{D_{Cu}}{D_{Ni}}$) was calculated and is tabulated in Table III with respect to equilibrium pH. The separation factor was the highest (123.8) at equilibrium pH 0.95 and the lowest (65.3) at equilibrium pH 0.56.

Extraction isotherms for copper and nickel and McCabe-Thiele plots

An aqueous solution containing 6.35 g/L copper(II) and 0.58 g/L nickel(II) with an initial pH of 3.95 was used for the extraction of copper with 20% (v/v) LIX®984N in kerosene. To determine the number of stages required for the extraction of copper, a McCabe-Thiele plot was constructed for O:A ratios from 1:5 to 5:1, with the total volume of the phases kept constant. The data in Figure 7 indicates that < 0.045 g/L copper will remain in the raffinate after two extraction stages at an O:A ratio of 3:4, which suggests that it could be possible to upgrade the copper concentration in the organic phase.

This prediction was confirmed by carrying out a two-stage batch countercurrent simulation study of up to five cycles at an O:A phase ratio of 3:4, in which the partially loaded organic from extraction stage 2 was fed to extraction stage 1. Analysis of copper and nickel values in the aqueous and organic phase confirmed the extraction of 8.4 g/L copper with negligible nickel co-extraction (0.005 g/L) in the loaded organic phase (Table IV). Thus, a loaded organic phase containing 8.4 g/L copper was generated.

Stripping of copper

The copper-loaded organic phase was stripped with different concentrations of sulphuric acid (5–25%) at equal phase ratios for 5 minutes. The aqueous phase was diluted and analysed for copper. The percentage copper stripping increased from 67.4% with 5% H$_2$SO$_4$ to 100% with 25% H$_2$SO$_4$. A McCabe-Thiele plot was constructed for the 20% loaded organic phase using 15% H$_2$SO$_4$ (98.2% stripping), with O:A phase ratios in the range 1:5 to 5:1. The data in Figure 8 suggests three stages of stripping at an O:A ratio of 3:1. A three-stage batch countercurrent experiment was carried out at the above O:A ratio using 15% H$_2$SO$_4$ strip

<table>
<thead>
<tr>
<th>Equilibrium pH</th>
<th>$D_{Cu}$</th>
<th>$D_{Ni}$</th>
<th>$\beta = \frac{D_{Cu}}{D_{Ni}}$</th>
</tr>
</thead>
<tbody>
<tr>
<td>1.08</td>
<td>11.43</td>
<td>0.10</td>
<td>119.92</td>
</tr>
<tr>
<td>0.95</td>
<td>10.87</td>
<td>0.09</td>
<td>123.8</td>
</tr>
<tr>
<td>0.91</td>
<td>9.85</td>
<td>0.09</td>
<td>113.61</td>
</tr>
<tr>
<td>0.92</td>
<td>7.99</td>
<td>0.09</td>
<td>102.05</td>
</tr>
<tr>
<td>0.76</td>
<td>6.06</td>
<td>0.07</td>
<td>84.0</td>
</tr>
<tr>
<td>0.7</td>
<td>5.24</td>
<td>0.07</td>
<td>74.85</td>
</tr>
<tr>
<td>0.56</td>
<td>4.18</td>
<td>0.07</td>
<td>63.3</td>
</tr>
<tr>
<td>0.36</td>
<td>1.57</td>
<td>0.01</td>
<td>114.81</td>
</tr>
</tbody>
</table>
Liquid-liquid extraction and separation of copper(II) and nickel(II) using LIX® 984N solution. The analysis of the aqueous phase showed an upgrading of copper to 25.1 g/L with 0.0089 g/L Ni in the loaded strip solution, leaving only 0.002 g/L copper in the organic phase (Table V).

Conclusions
The extraction of 0.1M Cu(II) from sulphate solution using 15% (v/v) LIX® 984N showed a pH dependence up to 0.61, after which it is independent of equilibrium pH. The extraction of copper increased with increasing extractant concentration. The species extracted into the organic phase is proposed to be CuR2. The equilibrium constant and Gibbs free energy were found to be 2.9 and 2.69 kJ mol⁻¹ respectively.

The percentage extraction of copper increased with increasing sodium sulphate concentration. With increasing dielectric constant of different diluents, the copper extraction was found to decrease. Effective separation of copper and nickel from an aqueous feed solution containing 6.35 g/L copper(II) and 0.58 g/L nickel(II) could be achieved with 20% (v/v) LIX® 984N in kerosene in the initial pH range (2.15–3.96), corresponding to an equilibrium pH of 0.91–1.08. A two-stage batch countercurrent extraction experiment at an O:A ratio of 3:4 resulted in upgrading the copper from 6.35 g/L in the feed solution to 8.4 g/L in the loaded organic phase.

Using 15% H₂SO₄ as the stripping agent, 25.1 g/L copper(II) was stripped into the loaded strip liquor, which could be suitable for electrowinning. A flow sheet is proposed for treatment of sulphate leach liquors obtained from pressure leaching of copper converter slags, nickel smelter slags, and copper sulphide concentrates.

Acknowledgements
The authors thank Cognis, Ireland Mining Chemical Technology Ltd, for providing the LIX® 984N. This research work was carried out with the encouragement and support of the authorities of Siksha’O’Anusandhan University.

References

Table IV
Results of a two-stage batch countercurrent simulation for copper extraction with 20% (v/v) LIX® 984N at an O:A ratio of 3:4

<table>
<thead>
<tr>
<th>Raffinate</th>
<th>Equilibrium pH</th>
<th>[Cu]₀₉₄ g/L</th>
<th>[Ni]₀₉₄ g/L</th>
<th>Organic phase</th>
<th>[Cu]₀₉₄ g/L</th>
<th>[Ni]₀₉₄ g/L</th>
</tr>
</thead>
<tbody>
<tr>
<td>R₂–5</td>
<td>0.87</td>
<td>0.03</td>
<td>0.583</td>
<td>E₁–5</td>
<td>7.56</td>
<td>0.003</td>
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<tr>
<td>R₂–4</td>
<td>0.9</td>
<td>0.029</td>
<td>0.585</td>
<td>E₁–5</td>
<td>8.24</td>
<td>0.0025</td>
</tr>
<tr>
<td>R₂–3</td>
<td>0.92</td>
<td>0.026</td>
<td>0.586</td>
<td>E₁–4</td>
<td>8.42</td>
<td>0.003</td>
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<tr>
<td>R₂–2</td>
<td>0.94</td>
<td>0.025</td>
<td>0.58</td>
<td>E₁–3</td>
<td>8.44</td>
<td>0.0026</td>
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<td>R₂–1</td>
<td>0.87</td>
<td>0.015</td>
<td>0.58</td>
<td>E₁–2</td>
<td>8.4</td>
<td>0.003</td>
</tr>
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</table>

Figure 8—McCabe-Thiele plot for stripping of loaded copper from 20% (v/v) LIX® 984N using 15%(v/v) H₂SO₄

Table V
Results of a three-stage batch countercurrent simulation for copper stripping with 15%(v/v) H₂SO₄ at an O:A ratio of 3:1

<table>
<thead>
<tr>
<th>Raffinate</th>
<th>[Cu]₀₉₄ g/L</th>
<th>[Ni]₀₉₄ g/L</th>
<th>Spent organic</th>
<th>[Cu]₀₉₄ g/L</th>
</tr>
</thead>
<tbody>
<tr>
<td>R₃–6</td>
<td>24.95</td>
<td>0.007</td>
<td>SO₁–6</td>
<td>0.003</td>
</tr>
<tr>
<td>R₃–5</td>
<td>25.1</td>
<td>0.0087</td>
<td>SO₁–5</td>
<td>0.002</td>
</tr>
<tr>
<td>R₃–4</td>
<td>25.0</td>
<td>0.0089</td>
<td>SO₁–4</td>
<td>0.0025</td>
</tr>
<tr>
<td>R₃–3</td>
<td>25.1</td>
<td>0.009</td>
<td>SO₁–3</td>
<td>0.002</td>
</tr>
<tr>
<td>R₃–2</td>
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<td>0.0089</td>
<td>SO₁–2</td>
<td>0.002</td>
</tr>
<tr>
<td>R₃–1</td>
<td>25.1</td>
<td>0.0089</td>
<td>SO₁–1</td>
<td>0.002</td>
</tr>
</tbody>
</table>

Liquid-liquid extraction and separation of copper(II) and nickel(II) using LIX® 984N


Figure 9—Proposed flow sheet for treatment of Cu-Ni leach liquor

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Conference Co-ordinator, Camiela Jardine
SAIMM, P O Box 61127, Marshalltown 2107
Tel: +27 11 834-127377
Fax: +27 11 833-8156 or +27 11 838-5923
E-mail: camiela@saimm.co.za
Website: http://www.saimm.co.za

BACKGROUND
Sub-Saharan Africa is endowed with vast mineral wealth, yet many of the region’s deposits have remained undeveloped. A key constraint is the lack of suitable transportation infrastructure from remote locations to the coast.

OBJECTIVES
The conference will provide a forum for key stakeholders to share perspectives on the opportunity, its challenges, and possible solutions.

CONFEERENCE
Accessing Africa’s Mineral Wealth: Mining Transport Infrastructure and Logistics
24–25 March 2015
Empress Palace Hotel Casino Convention Resort, Johannesburg