Introduction
The economic recovery of nickel from laterite ores has been pronounced for some years and will become more critical in the future, as lateritic ores constitute most of the world’s known nickel and cobalt resources, with nickel production from sulphide deposits progressively decreasing. High pressure acid leaching (HPAL) is being used for the recovery of nickel from nickel laterite ores, and increasingly atmospheric leaching is also being considered. The quantity of laterite resources amenable to hydrometallurgical processing (limonite, nontronite/smectite) is almost twice that amenable to pyrometallurgical processing (saprolite, garnierite) (Bacon and Mihaylov, 2002).

Various hydrometallurgical flow sheets are being used for the recovery of nickel from laterite ores. Most plants use the Caron or HPAL processes, for which simple block flow diagrams are given in Figure 1. The Caron process is considered primarily for limonitic ores to avoid the high acid consumption associated with the iron content in the ore. The ore is calcined reductively to reduce the ferric (associated with goethite), prior to ammonical leaching of the nickel and cobalt. Yabulu Nickel Refinery, Queensland, Australia implemented the Caron process.

The HPAL process can be used for limonitic as well as saprolitic ores (<4% Mg) and has been installed on numerous plants, including Moa Bay, Bulong, Murrin Murrin, and Goro (Bacon and Mihaylov, 2002). However, the actual recovery of nickel and cobalt and their separation primarily from calcium, magnesium, and manganese is done employing different flow sheets. In the Murrin Murrin flow sheet the pH of the HPAL pregnant solution is adjusted to pH 3.5–4 to neutralize excess acid and precipitate most ferric, aluminium, and chrome. This is followed by sulphide precipitation of nickel and cobalt, which is the primary technology employed to separate the nickel and cobalt from manganese, magnesium, and calcium.

The Bulong flow sheet (Figure 2) also neutralized the free acid and precipitated ferric, aluminium and chromium, but it employed direct solvent extraction (SX) for the recovery of nickel and cobalt (Flett, 2005). Cyanex 272 (2,4,4-trimethylpentyl phosphinic acid) was used to recover cobalt from the neutralized stream, followed by nickel SX using Versatic 10 acid (V10, a tertiary-branched carboxylic acid). One of the major drawbacks of direct SX as operated at Bulong was that the selectivity of the V10 extractant was inadequate to prevent calcium loading during extraction. This resulted in gypsum precipitation in the SX circuit, and hence major operational difficulties.

The most recent major nickel laterite project, namely Vale Inco’s Goro nickel project, was commissioned during 2012. Nickel is also recovered via direct SX (no prior precipitation

Synopsis
Mintek has been involved in extensive test work since the early 1990s on the recovery of nickel and cobalt from leach liquors saturated in calcium, using synergistic solvent extraction systems. During this period the Nicksyn™ reagent was developed, optimized, commercially manufactured, and tested by Tati Nickel on a demonstration plant for more than 2800 operating hours. Efficient recovery of nickel without the co-extraction of calcium, thus avoiding gypsum formation in the extraction and stripping circuits, was illustrated. This synergistic system was recently evaluated on a laboratory scale for the recovery of nickel and cobalt from synthetic lateritic sulphate leach liquor containing about 3 g/L nickel, 0.5 g/L cobalt, 0.7 g/L manganese, 20 g/L magnesium, and with calcium at saturation. Extraction and stripping parameters were determined for this feed liquor and are discussed in this paper.

Keywords
solvent extraction, nickel laterites, reagents, synergistic systems.

Evaluation of a Versatic 10 acid/Nicksyn™ synergistic system for the recovery of nickel and cobalt from a typical lateritic leach liquor
by A.C. du Preez* and M.H. Kotze*

Mintek has been involved in extensive test work since the early 1990s on the recovery of nickel and cobalt from leach liquors saturated in calcium, using synergistic solvent extraction systems. During this period the Nicksyn™ reagent was developed, optimized, commercially manufactured, and tested by Tati Nickel on a demonstration plant for more than 2800 operating hours. Efficient recovery of nickel without the co-extraction of calcium, thus avoiding gypsum formation in the extraction and stripping circuits, was illustrated. This synergistic system was recently evaluated on a laboratory scale for the recovery of nickel and cobalt from synthetic lateritic sulphate leach liquor containing about 3 g/L nickel, 0.5 g/L cobalt, 0.7 g/L manganese, 20 g/L magnesium, and with calcium at saturation. Extraction and stripping parameters were determined for this feed liquor and are discussed in this paper.

Keywords
solvent extraction, nickel laterites, reagents, synergistic systems.

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(i) The very strong extraction of copper requires efficient removal of copper from the full flow of the pregnant leach solution via ion exchange (IX), which would be expensive. Furthermore, if any breakthrough from the IX circuit occurs, the copper would report to the Cyanex 301 circuit, requiring the copper to be stripped with thiourea in sulphuric acid medium.

(ii) Strong nickel and cobalt extraction makes stripping difficult. The Goro process was designed to use hydrochloric acid stripping in four stripping stages, each with 5 minutes’ residence time in the mixer, at an operating temperature of 60°C and a residual hydrochloric acid concentration of 3 M.

(iii) Due to the high residual acid concentration required for stripping, pyrohydrolysis is used for nickel recovery as NiO.

(iv) The introduction of chloride into the system requires more sophisticated materials of construction, and can have environmental concerns among other complications.

(v) Cyanex 301 is chemically unstable and in the presence of oxygen and metals such as ferric, the reagent is oxidized to form a disulphide. Hence, air has to be excluded from the operating system, which Goro achieves by employing Bateman Pulsed Columns. The reagent can be regenerated using sulphuric acid and zinc powder.

Mintek developed the Nicksyn™ reagent during the 1990s, and together with V10 it provides an alternative and more cost-effective approach to direct nickel and cobalt SX from laterite processing liquors. This synergistic system has been demonstrated over 2800 hours on the Tati Nickel Activox® demonstration plant in Botswana, and has since been commercialized (Du Preez et al., 2007; Masiiwa et al., 2008). This paper describes the technical performance of the V10 and Nicksyn™ synergistic system for the recovery of nickel and cobalt from neutralized HPAL lateritic leach liquor.

Figure 1—Primary hydrometallurgical processing options for lateritic nickel ores (Dalvi, Bacon, and Osborne, 2004)

Figure 2—Simplified block flow diagram of the Bulong flow sheet

Figure 3—Metal extraction by 15 vol.% Cyanex 301 at an O:A phase ratio of 0.5 using NaOH for pH adjustment
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Experimental procedures

Laboratory tests

Analytical methods
All metal analyses were done by inductively coupled plasma-optical emission spectroscopy (ICP-OES) with a detection limit of 2 mg/L for all metals. Organic samples were stripped with sulphuric acid (approx. 1 M) at an organic-to-aqueous (O:A) phase ratio of 0.5, after which the strip liquors were submitted for analysis.

Reagents and solutions
Versatic 10 acid (V10, a tertiary-branched carboxylic acid) was obtained from Chemquest (produced by Resolution Performance Products Ltd.), while Nicksyn™ was prepared for Mintek by an international, reputable manufacturer. The chemical composition and technical information on Nicksyn™ remain the proprietary information of Mintek and can therefore not be disclosed. Appropriate dilutions of V10 alone and V10 mixtures with Nicksyn™ were done using an aliphatic hydrocarbon diluent, Shellsol D70, which was obtained from Shell Chemicals.

Metal distribution studies
Metal distribution equilibria (pH vs. extraction profiles, extraction and stripping isotherms) were generated by contacting the required organic phase with the appropriate feed solution at various O:A phase ratios, using rapid magnetic stirring and controlling the temperature in a water-jacketed glass vessel at 25°C. An equilibrium time of 10 to 15 minutes was allowed to ensure steady state was reached.

The pH value of the aqueous phase (in the case of extraction isotherms) was adjusted or controlled by the addition of sodium hydroxide solution (approx. 1 to 10 M), using a calibrated combined glass reference electrode. Samples of the organic phase were taken immediately after the aqueous samples to prevent possible re-equilibration after each pH adjustment. Aqueous samples were submitted for analyses. Organic samples were stripped with 1 M H₂SO₄ (O:A phase ratio of 0.5), after which the strip liquors were analysed via ICP-OES.

Organic phase (0.5 M V10 plus 0.25 M Nicksyn™) was batch-loaded for stripping purposes by contacting portions of fresh organic phase with synthetic laterite solution (at an O:A phase ratio of 0.45) at pH 6.0 and at 25°C. Samples of aqueous and organic phases were analysed by ICP-OES. This procedure was repeated two more times to simulate the three stages required according to the McCabe-Thiele construction (Figure 7). The batch-loaded organic phase obtained was then contacted with a synthetic nickel spent electrolyte (approx. 71 g/L nickel in 40 g/L H₂SO₄) at different O:A phase ratios at 25°C, measuring the final pH values of the loaded strip liquors. Samples of the loaded strip liquors and organic phases were analysed as previously described.

For the batch countercurrent extraction experiment, organic and aqueous phases were contacted (at 25°C), using magnetic stirring at an O:A phase ratio of 0.45. A sequence of batch contacts that simulates the conditions of a four-stage continuous flow process was used as illustrated in Figure 4. Six full cycles (D to I, see Figure 4) were completed in this way to ensure steady-state conditions. Samples of the raffinates of the fourth stages (4D to 4I), and the aqueous phases of the first (1I), second (2I), and third stage (3I) of the last cycle (I) were submitted for analyses. Portions of the loaded organic phases of the first stages (1D to 1I) as well as the loaded organic phases of the second (2I), third (3I) and fourth (4I) stages of the last cycle were taken and stripped as described above, after which the strip liquors were analysed by ICP-OES.

Results and discussion

Direct recovery of nickel and cobalt from a synthetic solution representing a nickel HPAL laterite leach liquor after iron, aluminium, and chromium removal was tested using the V10/Nicksyn™ system. Nicksyn™ is now commercially available, and hence offers a very attractive option to be considered for HPAL leach liquor.

Feed solution
A synthetic feed solution was made up from metal sulphate salts to contain nickel, cobalt, manganese, calcium (at saturation), and magnesium. The average of various analyses of the synthetic laterite feed solution is given in Table I.

<table>
<thead>
<tr>
<th>Feed</th>
<th>Concentration, g/L</th>
</tr>
</thead>
<tbody>
<tr>
<td>Ni</td>
<td>3</td>
</tr>
<tr>
<td>Co</td>
<td>0.5</td>
</tr>
<tr>
<td>Mn</td>
<td>0.66</td>
</tr>
<tr>
<td>Mg</td>
<td>20.2</td>
</tr>
<tr>
<td>Ca</td>
<td>0.46</td>
</tr>
</tbody>
</table>

Table I

Average composition of synthetic laterite feed solution

Figure 4—Scheme of contacts for batch countercurrent extraction experiment
Evaluation of a Versatic 10 acid/Nicksyn™ synergistic system for the recovery of nickel

Organic phase compositions

The different concentrations of V10 and the molar ratios of V10:Nicksyn™ diluted in Shellsol D70 are given in Table II.

Extraction metal-distribution equilibria (pH vs. extraction)

The origin of the synergistic effect for nickel by a carboxylic acid (such as V10, which exists in the form of dimers H₂A₂), with the addition of a synergistic compound (L) such as Nicksyn™, has been discussed previously in terms of competing equilibria, and is given in Equations 1 to 3 (Du Preez et al., 2007; Masiiwa et al., 2008):

\[
Ni^{2+} + H_2A_2 + 2L = NiA_2L_2 + 2H^+ \quad [1]
\]

\[
Ca^{2+} + 3H_2A_2 = CaA_3{(HA)}_3 + 2H^+ \quad [2]
\]

\[
H_2A_2 + 2L = 2HAL \quad [3]
\]

where \(H_2A_2\) denotes the carboxylic acid dimer and \(L\) denotes the synergist.

Results for selected pH vs. extraction isotherms are shown in Figure 5 and Figure 6. The pH₅₀ values (the pH at which 50% of the metal originally present in the aqueous phase is extracted under a given set of conditions) are summarized in Table III.

Synergistic shifts in the pH₅₀ values for the extraction of nickel (i.e. the difference in pH₅₀ value for V10 alone and the pH₅₀ value for the appropriate synergistic mixture) increased from 1.23 to 1.68 units when Nicksyn™ addition was increased from 0.125 to 0.5 M, whilst the shifts for cobalt increased from 0.65 to 1.15 units with the same Nicksyn™ additions. The extraction of manganese was largely unaffected, hence the separation (ΔpH₅₀Mn-Co) between cobalt and manganese increased from 0.75 to 1.13. Extractions of calcium and magnesium were negligible (<1%) under these conditions. This synergistic system therefore not only provides an option for the recovery and separation of nickel and cobalt efficiently from calcium and magnesium, but also gives the option of selecting a degree of manganese removal, with ease of pH control in practical flow sheets.

Extraction isotherms

The distribution isotherms and McCabe-Thiele constructions for the extraction of nickel and cobalt from synthetic laterite leach solution generated using 0.5 M V10 plus 0.25 M Nicksyn™ in Shellsol D70 at 25ºC.

### Table II

<table>
<thead>
<tr>
<th>V10</th>
<th>Nicksyn™</th>
<th>V10:Nicksyn™</th>
</tr>
</thead>
<tbody>
<tr>
<td>Vol.%</td>
<td>M</td>
<td>M</td>
</tr>
<tr>
<td>9.6</td>
<td>0.50</td>
<td>-</td>
</tr>
<tr>
<td>9.6</td>
<td>0.50</td>
<td>0.125</td>
</tr>
<tr>
<td>9.6</td>
<td>0.50</td>
<td>0.25</td>
</tr>
<tr>
<td>9.6</td>
<td>0.50</td>
<td>0.50</td>
</tr>
</tbody>
</table>

### Table III

<table>
<thead>
<tr>
<th>V10, M</th>
<th>Nicksyn™, M</th>
<th>pH₅₀ Mg</th>
<th>pH₅₀ Ca</th>
<th>pH₅₀ Mn</th>
<th>pH₅₀ Co</th>
<th>pH₅₀ Ni</th>
<th>pH₅₀ Mn-Co</th>
<th>pH₅₀ Ca-Co</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.5</td>
<td>-</td>
<td>&gt;7.17</td>
<td>&gt;7.17</td>
<td>&gt;7.17</td>
<td>6.52</td>
<td>6.37</td>
<td>&gt;0.65</td>
<td>&gt;0.65</td>
</tr>
<tr>
<td>0.5</td>
<td>0.125</td>
<td>&gt;7.19</td>
<td>&gt;7.19</td>
<td>&gt;7.19</td>
<td>6.62</td>
<td>5.87</td>
<td>5.14</td>
<td>0.75</td>
</tr>
<tr>
<td>0.5</td>
<td>0.25</td>
<td>&gt;7.43</td>
<td>&gt;7.43</td>
<td>&gt;7.43</td>
<td>6.47</td>
<td>5.53</td>
<td>4.97</td>
<td>&gt;1.90</td>
</tr>
<tr>
<td>0.5</td>
<td>0.5</td>
<td>&gt;7.30</td>
<td>&gt;7.30</td>
<td>&gt;7.30</td>
<td>6.50</td>
<td>5.37</td>
<td>4.69</td>
<td>1.13</td>
</tr>
</tbody>
</table>

* A > sign preceding a pH value indicates that 50% metal extraction was not reached at this pH value
Nicksyn™ (at pH 6.0) and 0.5 M V10 plus 0.5 M Nicksyn™ (at pH 5.8) in Shellsol D70 at 25ºC are shown in Figure 7 and Figure 8, respectively. The McCabe-Thiele construction was drawn for optimum recovery of nickel with the idea to gauge what the possible recovery for cobalt could be under the chosen conditions. In order to determine the effect of insufficient aluminium removal prior to the SX circuit, about 200 mg/L aluminium (as sulphate) was added to the leach solution used for the generation of the extraction isotherms.

The McCabe-Thiele construction on the isotherm indicated that a loading of about 7.1 g/L nickel could be achieved in three countercurrent extraction stages at an O:A phase ratio of 0.45. The maximum loading of cobalt under these conditions was about 300 mg/L. Calcium and manganese co-extraction were about 6 mg/L and 34 mg/L, respectively.

The results for the McCabe-Thiele construction shown in Figure 8 indicated that a slightly higher loading of about 9.6 g/L nickel could be achieved with the increased ratio of V10:Nicksyn™ of 1:1 in three countercurrent extraction stages at an O:A phase ratio of 0.33. Cobalt loading under these conditions was slightly lower than 400 mg/L. The co-extractions of calcium and manganese under these conditions were about 4 mg/L each.

Although slightly better separation (1.13 vs. 0.94 pH units) between cobalt and manganese, and higher loading of cobalt together with nickel (7.2 vs. 9.7 g/L) were achieved using a V10:Nicksyn™ ratio of 1:1 compared with the V10:Nicksyn™ ratio of 2:1, respectively, the additional costs of the increased Nicksyn™ concentration should be considered on an economic basis for each application.

In order to recover all cobalt together with nickel, the O:A phase ratio (and possibly the number of stages) would have to be adjusted to compensate for cobalt being ‘crowded off’ as illustrated in the unfavourable isotherms obtained for cobalt (Figure 9 and Figure 10).

The McCabe-Thiele construction redrawn for optimum cobalt recovery under the conditions tested indicated that a higher O:A phase ratio (1.86 vs. 0.45 as previously drawn for nickel recovery) should be employed to ensure a loading of approximately 257 mg/L cobalt without being ‘crowded off’ by nickel, using two countercurrent extraction stages. Under these conditions, nickel would still be recovered (leaving <2 mg/L in the raffinate) with minimum co-loading of calcium (5 mg/L), magnesium (4 mg/L), and manganese (7 mg/L).

For optimum cobalt recovery under these conditions the McCabe-Thiele construction indicated that an O:A phase ratio of 1.27 (instead of 0.33 previously drawn for nickel recovery)
would be required to achieve a loading of about 400 mg/L cobalt without ‘crowding off’ by nickel in two to three countercurrent extraction stages. Under these conditions, a similar nickel recovery would still be expected (<2 mg/L in the raffinate) with minimum co-loading of calcium (4 mg/L), magnesium (4 mg/L), and manganese (14 mg/L). In both these cases optimum design should include economic cobalt recovery without calcium co-extraction.

**Batch countercurrent test work**

A batch countercurrent test was performed for the extraction of cobalt and nickel with 0.5 M V10 plus 0.25 M Nicksyn™ in Shellsol D70 at 25°C. Four extraction stages were used with an O:A phase ratio of 0.45 and an equilibrium pH value of 6.0 in each stage (i.e. a flat pH profile) over a total of six full cycles (see Figure 4). The results are illustrated in Figure 11.

A loaded organic phase (Stage 1%) containing about 7.7 g/L nickel was obtained at steady state (Cycle 1, Figure 4), leaving about 5 mg/L nickel in the raffinate, which related to >99% extraction. Cobalt was loaded up to approximately 1.4 g/L in stage 3I, after which it was ‘crowded off’ by nickel to only about 176 mg/L on the loaded organic phase (Stage 1%) under these conditions. In order to recover both cobalt and nickel, a higher O:A phase ratio and probably more stages would be required, as shown in Figure 9 and Figure 10. A pH profile (and not a flat profile as was employed here) over all the stages could also assist to enhance cobalt recovery, providing no calcium is co-extracted.

The co-loading of some impurities at steady state of the batch countercurrent experiment is shown in Figure 12.

Further optimization of the O:A phase ratio, number of stages, and pH profile across the extraction circuit would have to be done to optimize cobalt recovery (including nickel) as well as limiting the co-loading of unwanted impurities such as manganese. This test work has to be performed for each individual lateritic feed solution.
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Stripping test work
In order to conduct stripping test work, a batch of fresh organic phase (0.5 M V10 plus 0.25 M Nicksyn™ in Shellsol D70) was preloaded to reasonably represent a loaded organic phase that would be expected from an extraction circuit. Synthetic spent nickel electrolyte was prepared to contain approximately 71 g/L nickel and 40 g/L H₂SO₄ (which should relate to a delta of approximately 24 g/L nickel). A stripping isotherm was generated and the results are shown in Figure 13.

Loaded strip liquor containing about 95 g/L nickel (at a pH value between 1.1 and 4.2) could be generated. The McCabe-Thiele construction indicated that two to three stages and an O:A phase ratio of 5.3 would be required to achieve this. Strip liquor pH values measured for stripping done at O:A phase ratios of 0.1 to 2.0 varied between 0.5 and 1.1, which indicates that adequate sulphuric acid was available for complete stripping of nickel. Stripping at O:A phase ratios of 5.0, 8.0 and 10.0 resulted in strip liquors exhibiting pH values of 4.2, 4.6, and 4.8 respectively, which indicated unfavourable stripping conditions for nickel. If plant operation required these operating conditions, it would be advisable to employ pH control (at about 3) in order to facilitate efficient stripping of nickel in a minimum number of stages and to provide an advanced electrolyte suitable for electrowinning.

Any co-loaded manganese could be removed by scrubbing with pH-adjusted water or a portion of the loaded strip liquor, depending on downstream requirements. The addition of a washing stage is recommended for removal of entrained aqueous phase from the loaded organic phase.

Conclusions
➤ The V10 plus Nicksyn™ synergistic mixture was evaluated for the use of direct solvent extraction of nickel and cobalt from lateritic leach liquor
➤ Nicksyn™ is now commercially available via a secured, reputable international supplier. It is a very attractive option and should be included in the evaluation for all HPAL hydrometallurgical projects
➤ For a solution containing about 3 g/L nickel, 0.5 g/L cobalt, 0.7 g/L manganese, 20 g/L magnesium, and 40 g/L H₂SO₄, the V10 plus Nicksyn™ synergistic mixture was found to be stable over a testing period of 100 days (representing stripping conditions) in the laboratory. The solubility was found to be between 1 and 3 mg/L under specific conditions tested (Du Preez and Preston, 2004).
➤ Mintek is currently involved in an international project with a commercial client where Nicksyn™ will be included in the definitive feasibility study for nickel and cobalt recovery from laterite solutions.

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References