

RECOVERY OF *p*-TBC FROM A BUTADIENE WASHING STREAM IN A PILOT PLANT

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Abstract - Results obtained in a pilot-scale unit designed for COD removal and *p*-TBC (*p*-tert-butylcatechol) recovery from a butadiene washing stream (pH 14, 200,000 mg COD L⁻¹, highly toxic) at a petrochemical industry are presented. By adding H₃PO₄, phase separation is achieved and *p*-TBC is successfully recovered (88 g L⁻¹ of washing stream). Information (time for phase separation and organic phase characterization) was gathered for designing a future industrial unit. The estimated heat generation rate was 990 kJ min⁻¹ and 15 min were enough to promote phase separation for a liquid column of approximately 1.15 m.

Keywords: Butadiene; *p*-TBC; Pilot plant; Industrial effluent.

INTRODUCTION

Butadiene is an important raw material within the petrochemical industry. It is used for producing polymers, copolymers, resins, and chemical intermediates. Unfortunately, its high reactivity may lead to undesirable reactions during manufacturing, storage, handling, distribution, and use, with butadiene peroxide formation and its own polymerization posing significant safety and health hazards.

To prevent those reactions from happening, butadiene is sold containing 120 to 150 mg kg⁻¹ of *p*-TBC (*p*-tert-butylcatechol), a polymerization inhibitor. However, several applications require *p*-TBC to be removed prior to butadiene use.

This work was performed at Petroflex Company, the largest synthetic rubber producer in Latin America, where *p*-TBC is removed from butadiene by washing it with a 15% w/w sodium hydroxide solution, although *p*-TBC could also be removed by

activated alumina adsorption (Rivero et al., 2003). This washing generates a highly toxic stream due to its high pH (around 14) and high *p*-TBC content (Reis et al., 2007), impairing the existing Industrial Wastewater Treatment Plant (IWTP) performance.

Phenolic compounds, like *p*-TBC, are valuable enough for drawing attention to procedures designed to separate and recover them from industrial streams (Venkateswaran and Palanivelu, 2006). Among them, several deal with solubilization with sodium hydroxide (separation) and further acidification (recovery). This approach has been reported by several authors (Han et al., 2001; González-Muñoz et al., 2003; Lazarova and Boyadzhieva, 2004; Ferreira et al., 2005; Jaber et al., 2005; Sawai et al., 2005).

In a previous work (Lóss et al., 2006), it was shown that acidification of Petroflex's butadiene washing stream with phosphoric acid was effective for Chemical Oxygen Demand (COD) removal and *p*-TBC recovery. An initial temperature of 60°C and

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an acidification down to pH 6 were necessary to avoid crystallization of salts after cooling. Results indicated a COD removal of 97% and a *p*-TBC recovery of 88 g L⁻¹ of washing stream. Since Petroflex generates approximately 15,000 L of that stream per month, that would mean a monthly recovery of 1,320 kg of *p*-TBC. Therefore, a significant reduction in the amount of organic matter entering the wastewater treatment unit is achieved.

There are two other important features: (i) the obtained COD removal would mean a reduction of 2,910 kg of oxygen per month in the wastewater treatment unit aerator, with the corresponding cost savings; and (ii) approximately 3,200 kg of CO₂ per month would cease to be emitted, making the treatment process greener.

Those figures motivated the implementation of a pilot-scale unit at the industrial site. Therefore, the present work reports the results obtained with the pilot-scale unit. The main objectives of the study were to obtain the necessary data to design a recovery process, more specifically three main parameters: heat release, the time necessary for phase separation after completing acidification, and the characterization of the organic phase. Although sulfuric acid has a lower cost compared to phosphoric acid, the latter was used for several reasons: (i) it is safer; (ii) allows better temperature control; (iii) there is no sulphate addition to the stream (avoiding future problems related to water

reuse); and (iv) Petroflex already adds polyphosphates to its wastewater streams as a nutrient (cost reduction).

EXPERIMENTAL

In this work, a butadiene washing stream from a rubber industry (Petroflex, Brazil), generated in the reaction area known as “Soda 23” (pH = 14, COD = 200,000 mg O₂ L⁻¹), was used. Concentrated phosphoric acid (85% w/w, Fosbrasil, Brazil) was used as received.

Figure 1 presents a scheme of the pilot plant. It was comprised of a 1 m³ 304 stainless steel insulated tank (2:1 semi-elliptic domes) equipped with: an impeller (pneumatic inclined turbine, BOMAX, Agimax F-170M-OP007R-400E1, Brazil), a temperature indicator (TI), and a level indicator (LI). “Soda 23” stream was transferred to the tank by a diaphragm pump (WILDEN, P1/PPP/WF/ WF/KWF, USA). Phosphoric acid was added by a dosing pump (OMEL, DMD 1, Brazil).

First, 600 L of “Soda 23” was fed to the tank followed by turning on agitation. After vapor-heating the tank up to 60°C, the dosing of phosphoric acid was initiated at a flow rate of 0.8 L min⁻¹. After the acid dosing, the agitation was ceased and samples withdrawn from the aqueous phase at regular time intervals.

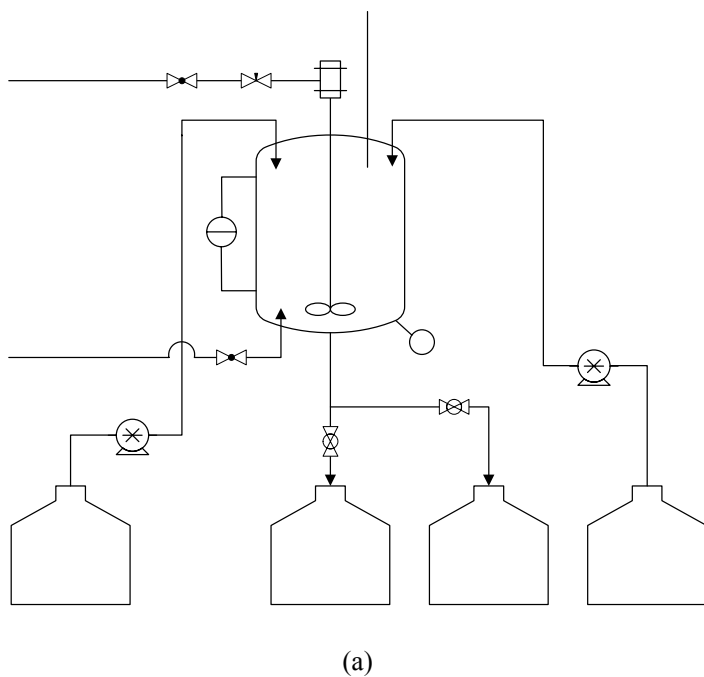


Figure 1: Pilot plant configuration: (a) schematic diagram and (b) photo of the pilot unit.

For determining the chemical oxygen demand (COD) of the aqueous phase, the closed-reflux method was used (APHA, 1992).

Chromatographic analyses were performed in order to determine the purity of the phase separated upon acidification. Prior to chromatography, samples were submitted to an adapted liquid-liquid extraction procedure using dichloromethane as the solvent (Czuczwa et al., 1987; Goldberg and Weiner, 1980).

A Shimadzu GC-17A gas chromatograph provided with a DB-5 (30 m × 0.25 mm × 0.25 μm) column and a FID detector was employed. The GC temperature was programmed: the injection and detector ports were held at 250°C and 280°C, respectively, and the GC oven temperature started at 75°C and was increased to 250°C at 10°C min⁻¹. The column head pressure was also programmed: it started at 40 kPa and was increased to 70 kPa at 1.7 kPa min⁻¹. The carrier gas (H₂) was then kept at a flow rate of 0.80 mL min⁻¹ and a linear speed of 25 cm s⁻¹. The extract (1.0 μL) was injected in split mode (1:30).

RESULTS AND DISCUSSION

The acidification of the wastewater in the pilot plant promoted phase separation. No salt

crystallization was observed after cooling down to ambient temperature.

Figure 2 shows two illustrative chromatograms. In Figure 2a, the chromatogram indicates that the “Soda 23” stream is composed of in fact two compounds: the one having a retention time of 11.310 min is *p*-TBC itself; the other at 8.829 min is probably due to a common impurity of commercial *p*-TBC, known as 3,5-di-tert-butylcatechol. However, as depicted by Figure 2b, the organic phase separated after acidification is solely composed of *p*-TBC. Therefore, it can be observed that no oxidation products are formed during the acidification and, in addition, that the recovered *p*-TBC is higher in purity than the commercial product.

The heat released during the neutralization of “Soda 23” increased the system temperature, as the tank was not equipped with a heat exchanger. Figure 3 shows the temperature increase during the experiments performed with a flow rate of 0.8 L min⁻¹.

As can be observed in Figure 3, temperature rises linearly up to approximately 96°C — 1st stage of the process — and then tends to stabilize — 2nd stage of the process.

During the first stage, the temperature rise is proportional to the mass of acid added to the system, as depicted also in Figure 3. The temperature increase rate was calculated as being 0.4 K kg⁻¹ of H₃PO₄.

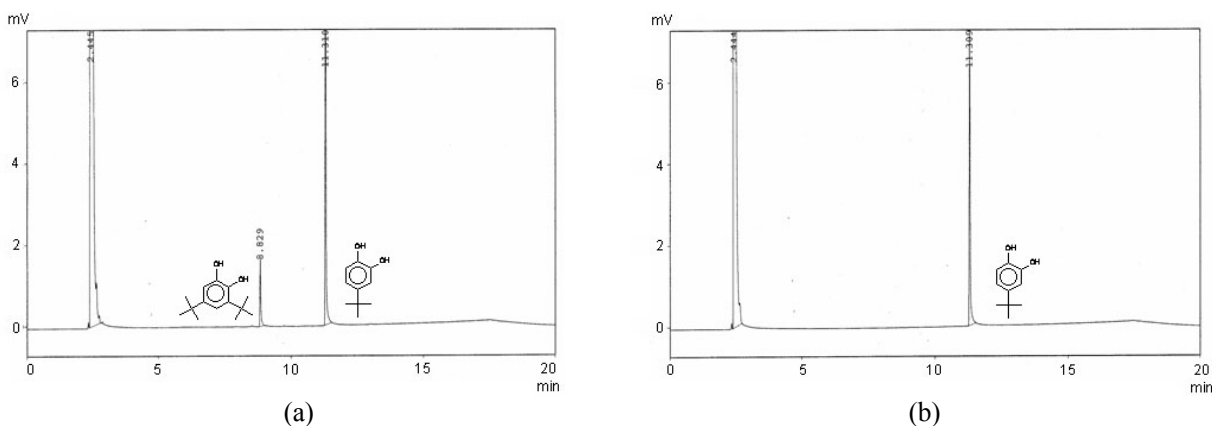


Figure 2: Chromatogram of (a) Soda 23 and (b) organic layer after acidification.

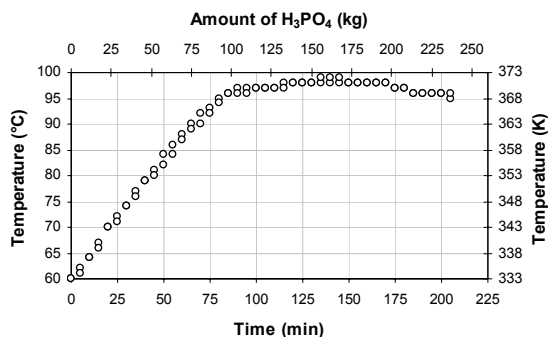


Figure 3: Temperature increase versus time and mass of acid added during Soda 23 acidification with H₃PO₄ at an acid flow rate of 0.8 L min⁻¹.

The temperature rise during the first stage is due to the heat released by the neutralization of sodium hydroxide, whose initial concentration is approximately 4.35 mol L^{-1} . Soda reacts with phosphoric acid up to a concentration less than 1.0 mol L^{-1} without any noticeable pH modification, and it is almost completely neutralized when the pH reaches 13. Subsequent additions of acid significantly modify the medium pH and the concentrations of anions (PO_4^{3-} , HPO_4^{2-} , and H_2PO_4^-) and of phosphoric acid itself. Nevertheless, those reactions are not exothermic (2nd stage of the process).

The theoretical mass of acid needed to acidify 600 L of wastewater to pH 13 is 88 kg. During the tests, the temperature stabilized when 100 kg of acid were added. Considering that an error of 1% in the acid flow rate represents approximately 11.5 kg during an addition period of 80 min (average time for the stabilization of the temperature), the real and theoretical acid consumption are close.

If a heat exchanger is to be designed for the system, it is necessary to know the heat exchange rate (Q_{ex}) of the process. As the reactor used was operated in a semi-batch mode, then the energy balance takes the form presented in Equation 1, where: \dot{m} is the feeding flow rate (kg min^{-1}), which is constant; m_0 and m are the initial and the total mass (kg) inside the tank, respectively; c_p and $c_p^{\text{H}_3\text{PO}_4}$ are the average heat capacity of the liquid inside the tank and that of the phosphoric acid ($\text{kJ kg}^{-1} \text{K}^{-1}$), respectively; T and T_f are the temperatures of the liquid inside the tank and that of the feeding one, respectively; ΔH is the reaction enthalpy (kJ mol^{-1}); r is the reaction specific rate ($\text{mol m}^{-3} \text{min}^{-1}$); V is the volume of liquid inside the reactor (m^3); and Q_1 represents the heat losses to the surroundings (kJ min^{-1}), which can be neglected because the vessel was insulated.

$$m \cdot c_p \cdot \frac{dT}{dt} = \dot{m} \cdot c_p^{\text{H}_3\text{PO}_4} \cdot (T_f - T) + (-\Delta H) \cdot r \cdot V + Q_1 \quad (1)$$

First, the feeding flow rate can be equaled to the acid mass flow rate, as losses from evaporation can be neglected. As $\rho_{\text{H}_3\text{PO}_4} = 1.685 \text{ kg L}^{-1}$ and $Q_{\text{H}_3\text{PO}_4} = 0.8 \text{ L min}^{-1}$, then $\dot{m} = 1.348 \text{ kg min}^{-1}$. The total mass inside the tank, at any time, can be calculate by Equation 2, which comes from the total mass balance, $dm/dt = \dot{m}$, and the corresponding integration assuming that \dot{m} is constant and that $m|_{t=0} = m_0$ (600 kg).

$$m = m_0 + \dot{m} \cdot t \quad (2)$$

Second, as depicted by Figure 3, the temperature variation during acid addition is constant (a straight line is observed); this is an indication that the specific heat capacity remains approximately unchanged. In order to calculate the specific heat capacity of the initial solution, the following steps were taken, while $c_p^{\text{H}_3\text{PO}_4}$ was retrieved from the literature as $1.53 \text{ kJ kg}^{-1} \text{K}^{-1}$ (NIST, 2008):

(a) The specific heat capacities (c_p) at 60°C for *p*-TBC, sodium hydroxide, and water were retrieved from literature: 0.2703 (Steele et al., 1997), 1.4178 (Cheresources, 2007), and 4.1843 (CRC, 2004) $\text{J g}^{-1} \text{K}^{-1}$, respectively;

(b) The mass percentages of each component were calculated (0.0905, 0.174, and 0.736 for *p*-TBC, sodium hydroxide, and water, respectively); and

(c) The solution specific heat capacity was estimated by a previously proposed method (Dimoplou, 1972), as presented by Equation 3.

$$c_p = 0.0905 \times 0.2703 + 0.174 \times 1.4178 + 0.736 \times 4.1843 = 3.3487 \text{ kJ kg}^{-1} \text{K}^{-1} \quad (3)$$

During the 1st stage of the process, it can be considered that the acid reacts instantaneously with the base present inside the tank, yielding Equation 4, where M is the molar mass (kg mol^{-1}) of the acid.

$$r \cdot V = \frac{\dot{m}}{M} \quad (4)$$

The substitution of Equation 4 into Equation 1 yields Equation 5, which can be integrated considering that $T = T_0$ at $t = 0$, resulting in Equation 6.

$$(m_0 + \dot{m} \cdot t) \cdot c_p \cdot \frac{dT}{dt} = \dot{m} \cdot c_p^{\text{H}_3\text{PO}_4} \cdot (T_f - T) + (-\Delta H) \cdot \frac{\dot{m}}{M} \quad (5)$$

$$T = \left(T_f - \frac{\Delta H}{c_p^{\text{H}_3\text{PO}_4} \cdot M} \right) - \frac{\left(T_f - \frac{\Delta H}{c_p^{\text{H}_3\text{PO}_4} \cdot M} - T_0 \right)}{\left(1 + \frac{\dot{m}}{m_0} \cdot t \right) \frac{c_p^{\text{H}_3\text{PO}_4}}{c_p}} \quad (6)$$

Then, when Equation 6 is adjusted to the 1st stage data of Figure 3, it is obtained that

$$T_f - \frac{\Delta H}{c_p^{\text{H}_3\text{PO}_4} \cdot M} \cong 830 \text{ K with a } R^2 = 0.999.$$

Finally, the required heat exchange rate in order to keep the liquid inside the tank at a constant temperature equal to the initial one, T_0 , can be calculated directly with Equation 1 replacing Q_l by Q_{ex} . In this case, $dT/dt = 0$ and Equation 7 is obtained. Therefore, the calculated Q_{ex} is approximately 990 kJ min^{-1} .

$$Q_{\text{ex}} = -\dot{m} \cdot c_p^{\text{H}_3\text{PO}_4} \cdot \left(-T_0 + T_f - \frac{\Delta H}{c_p^{\text{H}_3\text{PO}_4} \cdot M} \right) \quad (7)$$

Phase separation was assessed by measuring the residual COD in the aqueous phase. Figure 4 shows the COD variation versus time after agitation was ceased (along with the 95% confidence limits), resulting from triplicate experiments, for an acid flow rate of 0.8 L min^{-1} . Initially, COD varies greatly but, from 15 min on, it is practically stabilized, indicating that phase separation was accomplished. This result is quite promising, as a 15-min separation time for a liquid column of approximately 1.15 m can be considered to be fast.

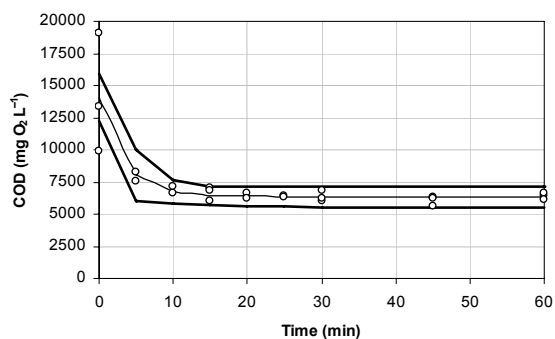


Figure 4: Aqueous phase COD variation versus time after acid addition (phase separation).

CONCLUSION

The recovery of *p*-tert-butylcatechol (*p*-TBC) from a butadiene washing stream at a petrochemical industry was assessed in order to obtain data for design purposes.

The neutralization of the stream with H_3PO_4 proved to be an excellent method, due to several reasons: (i) no oxidation products are formed during neutralization; (ii) the recovered *p*-TBC has a higher

purity than the commercial *p*-TBC used; and (iii) the time needed for complete phase separation after acid addition is quite short (15 min for a liquid column of 1.15 m).

For designing a future heat exchanger for the recovery process, the heat generation rate was calculated as being 990 kJ min^{-1} during the acid addition step, for an acid flow rate of 0.8 L min^{-1} .

It is estimated that a monthly recovery of 1,320 kg of *p*-TBC can be achieved, decreasing significantly the organic load in the effluent to the biological process used. Cost savings are also significant due to the use of the recovered *p*-TBC and to the reduced oxygen consumption in the aeration tank (2,910 kg per month). Finally, it was also estimated that the recovery process would prevent 3,200 kg of CO_2 from being emitted to the atmosphere per month, making the industry operation greener.

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